



# MASSIVE HYDRAULIC FRACTURING EXPERIMENTS OF THE DEVONIAN SHALE IN LINCOLN COUNTY, WEST VIRGINIA

COLUMBIA/DOE CONTRACT E (46-1)-8014

## APPENDICES

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TERRA TEK REPORT

IN SITU STRESS DETERMINATION IN THE DEVONIAN SHALE  
(IRA MCCOY 20402) WITHIN THE ROME BASIN

VOLUME II

APPENDICES

APPENDIX I



APPENDIX 1: AN EXPLANATION OF GAS CONTENT TERMINOLOGY  
AND ANALYTICAL METHODS

Standardization of Reserve Terminology

Considering the importance of the reported gas reserve estimates to exploration planning, caution should be exercised when referring to any of the reported values, especially when such figures might be subject to public scrutiny. At some future time, we shall be called to account for the gas reserves which have been published. If it is generally known that some given quantity of gas is contained in the Eastern Devonian Shales, it may be difficult to explain after the fact that only a fraction of the reserves are exploitable. With this in mind, it seems logical to select those figures which most closely reflect the economically producible gas reserves.

The most reliable information is the "produced gas" as given by actual production records, but this can only be determined after a well has ceased production. However, well log and reservoir test data are used to estimate the amount of "free gas in place" and "recoverable gas," and with additional production data one may accurately calculate the recoverable gas reserves. The fraction of the gas in place which is recoverable will depend on the back pressure in the gas lines and technical and economic factors. Perhaps the best reserves figures are based on either produced or producible gas values, but in order to use these estimates to determine the reserves elsewhere, we must assume a drainage volume for each well, make the assumption that the reservoir contains a similar gas concentration at every location and the stimulation method is effective. Such assumptions will introduce an undetermined amount of error.

In the present attempts to calculate reserves for the entire Appalachian Basin from the analyses of canned rock samples, it would be most desirable to base the calculations on some reference pressure, as we cannot continue to

produce below the line back pressure. Columbia's estimates relate the gas content to the pressure increase observed in sealed sample containers--that is, the final gauge pressure. Samples which contain some hydrocarbon gas but yield no measurable pressure increase are considered to contain no producible gas. Although the true producible portion will vary depending on the actual collection line pressure, it is felt that our values which we shall call "off gas" are the most applicable. Moreover, the term "off gas" doesn't imply any physical state, but merely that it is mobile gas of some sort. It should be noted that the "off gas" as measured does not include any gas originally present in fractures, or other permeable zones.

Battelle's "free gas" values are more difficult to apply since the gas pressure includes the initial air pressure of 14.7 psi. Even if these figures are multiplied by the hydrocarbon fraction of the contained gas, this will include gas which is not producible with a back pressure of zero psig. Battelle's "weight loss" values from thermogravimetric measurements are almost entirely due to water vapor release. Consequently, they are of no use to us. Our own low mobility hydrocarbon content is also of limited economic interest at this time.

"Gas yield," "oil yield," "Fischer Assay," "gas equivalent of carbon content" and related hydrocarbon content figures are not measures of the gas content of a formation; but are, in fact, the hydrocarbon fractions which may only be generated and liberated by heating rock volumes to a sufficient temperature under proper conditions. These figures must not be confused with gas contents or true gas reserves, as the economics of exploitation differ markedly. It is difficult to compare the gas yield and gas equivalent results from different laboratories due to differences in analytical procedures (temperatures, pressures, addition of hydrogen, and catalytic agents). The ultimate gas-equivalent reserves can only be determined by that retorting procedure which generates the maximum possible amount of gas.

In summary, in the absence of production or well test data, the most applicable type of rock or core gas measurement is one based on observed pressure increases due to hydrocarbon gas release as is being done by the Columbia Gas Research Department. It is proposed that this fraction of the total gas content be referred to as the "off gas" content. The adoption of some standard terminology appears to be a first step in working towards a basis of comparison for results, and a basin-wide extension of reserve estimates based on core analyses.

## Categories of Hydrocarbons and Other Organic Matter in Relation to Resource Evaluation

Anyone who reviews the existing literature on the hydrocarbon reserves of organic rich sediments will find a bewildering range of reported values. These differences in gas and oil content may be real and due to actual horizontal and vertical gradations, or may be only apparent and related to nomenclature differences, the inclusion of only one or several physical states of hydrocarbons in the rock, differences in methods of measurement or content calculation models. The following outline is an attempt to clarify some of this confusion.

### I. Nomenclature and Physical Forms of Hydrocarbons and Organic Compounds:

#### A. Light, hydrocarbons which are gases under standard conditions include free gas, adsorbed gas and trapped gas.

1. "Gas in place" is the total content of light hydrocarbons which would be gaseous under standard conditions. This corresponds to the maximum recoverable volume if economic factors are disregarded, but below the temperatures at which kerogen decomposes. (Includes all of the other forms: free, adsorbed, trapped, fracture content, etc.).
2. "Produced gas" is that volume of gas actually recovered according to production records.
3. "Recoverable gas" is that calculated portion of the total gas in place which can be exploited economically against a given line pressure.
4. "Off gas" refers to those lighter hydrocarbons which can escape from a rock volume under ambient conditions given sufficient time without any heat stimulation and without grinding the rock. This term does not imply that the gas is in any one physical state in situ.
5. "Fracture gas" is that gas which is contained in the fracture volume. (Cannot be accurately measured at present).
6. "Adsorbed gas" also includes the gas which will eventually be released without grinding or heat treatment, but may include trapped gas which is totally enclosed by mineral grains and cement. (Research conducted at Juniata College suggests that what has been called free gas may behave entirely as an adsorbed phase. The adsorption model allows for multiple layer concentrations which would have a density similar to a liquid).

7. Low Mobility hydrocarbons include that portion totally surrounded by mineral grains (trapped gas) and other heavy hydrocarbons which are only released at a noticeable rate at temperatures of about 300°C.
8. Heavy hydrocarbons exist as liquids under standard conditions.
  1. Oil and related compounds
- C. Solids
  1. Bitumens are solvent extractable.
  2. Kerogen is a residual macromolecular residue which is insoluble.

## II. Methods of Measuring Hydrocarbon Content

- A. Light hydrocarbons (gas components) may be measured by:
  1. Determining the volume and pressure of gases released at ambient temperature (off gas or gas in place).
  2. Determining the weight loss of a sample with a thermogravimetric balance at temperatures ranging up to 100°C as was done by Battelle mostly shows the amount of adsorbed water rather than the hydrocarbon gas content.
  3. Determining gas released by grinding samples in sealed containers. (trapped gas).
  4. Determining gas actually produced from a well assuming a given drainage volume. (producible gas or gas in place)
- B. Heavy hydrocarbons may be measured by:
  1. Retorting (oil yield)
  2. Solvent extraction (content of various fractions)
- C. Solids may be measured by:
  1. Retorting, distillation, pyrolysis (oil/gas yield--variable yields due to variable operating conditions. At temperatures near 300°C, kerogen appears to begin to break down).
  2. Solvent extraction (content of various fractions)

## GAS CONTENT CALCULATIONS

1. "Off gas" values determined by Columbia's core gas analyses from Lincoln County, West Virginia core samples (Method of measurement described previously in the test):

Range 0 to 1.8 scf/cf shale  
Mean 0.2 scf/cf shale

2. Values reported by Battelle and referred to by them as "free gas" from our Lincoln County samples (Includes the air sealed in the cans; therefore, the values are high):

Range 1.4 to 2.9 scf/cf shale

3. Rejected recoverable gas reported by Columbia by K. L. Brooks and R. M. Forrest, 1974, based on recorded production from the Big Sandy Field of 198 MMcf assumed drainage area of 40 acres and stimulated section thickness of 500 feet.

Calculated gas content = .23 scf/cf shale

4. "Adsorbed gas" from Lincoln County samples

- A. Measured by Battelle by TGA weight loss described subsequently. This is determined after the "off gas" has been removed. The weight loss now known to be due to water vapor desorption.

Range 18 to 50 scf/cf shale

- B. Juniata is progressing toward an "adsorbed gas" content calculation based on the gas released into the sample containers, but no values have been reported yet.

5. "Trapped gas" in shale cores as reported by Columbia is determined from the area under the 250°C peak on a plot of the total hydrocarbon gas released per unit time as 1-3 gram samples are heated from ambient temperature to 600°C.
6. Gas yield from retorting, distillation, or pyrolysis (Care should be exercised in comparing these results because of differences in the analytical techniques.)
  - A. Based on values in the literature (Ashley and other papers), and some experiments by K. Brooks whereby shale samples were heated to 300°C or more, Columbia has referred to a value of 1 Mcf/ton or

gas yield = 87 scf/cf shale

(Due to the lack of detail as to the exact temperatures and other related Parameters, the exact significance of these results is questionable. It is, therefore, difficult to compare these values, and better not to refer to them.)

- B. Based on Battelle's values reported for gas equivalent of the carbon content:

gas yield - from 50 to 360 scf/cf shale  
(All carbon is converted to gas.)

- C. Columbia's Kerogen-derived gas is measured by the area under the 450°C peak of a pyrolysis temperature plot similar to the trapped gas.

### Recoverable Gas Potential

Shale sequences are generally considered to be the principal loci of hydrocarbon generation and are known to contain certain volumes of natural gas. They are typically low in permeability. Due to the low permeability, shale gas production has been restricted to highly fractured zones where a network of interconnecting fractures can provide a suitable reservoir. The goal of our shale fracturing program is to create similar open fracture systems of sufficient surface area to permit a rapid transfer of gas from the shale bodies to the well bore. The formation porosity, formation permeability, actual gas content, and the mode of occurrence of the gas are of primary importance in determining the resource potential of a shale sequence. Porosity measures the maximum volume within the formation which could be occupied by free gas. Permeability determines the rate at which gas can move through the shale (from which we may estimate a theoretical gas production rate). From a knowledge of the gas content, we can assign a limit to the production we may expect from a given area or shale volume. A knowledge of the mode of gas occurrence (gaseous, liquid or adsorbed phase) will also be necessary to determine the equations for calculating the gas content.

Measurements of these properties have been performed on samples from Well 20155-T in Belmont County, Ohio and Well 20403-T in Lincoln County, West Virginia. The analytical work on the Belmont County samples was done at Battelle Memorial Institute, Columbus, Ohio and at the Chemistry Department of Juniata College, Huntingdon, Pennsylvania. Samples from the Lincoln County well have been analyzed at Core Laboratories, Dallas, Texas; Halliburton Services, Duncan, Oklahoma; and Columbia Gas System Service Corporation, Marble Cliff, Ohio. Schlumberger and Birdwell logs also contain porosity values for these wells.

A considerable range of values has been reported for the gas reserves of the eastern Devonian Shale. The U.S.G.S. claims yields of over 1,000 cubic



feet of gas may be produced by "distillation" of one ton of shale. Values as high as 30 cubic feet of natural gas per cubic foot of shale have been reported (Battelle), but other values as low as one cubic foot per ton are reported. In view of the importance of these reserve figures, it is highly desirable to determine the precision and accuracy of the data upon which we shall base our estimates. The porosity and gas content values reported for our Devonian Shale samples have shown a wide range of values which, in turn, would give rise to a large range in the calculated gas reserves. Although actual variations in the gas content of the shale are to be expected for different locations and depths, some apparent variation is related to differences in the methods used to determine gas content and porosity.

#### Methods:

##### Porosity Determination

The porosity values reported by Battelle were obtained by standard gravimetric techniques--bulk density, apparent density (from weight suspended in  $\text{CCl}_4$ ), and true density (from pycnometer measurements). From the three types of measurements, one may calculate the open, closed, and total porosities. There appear to be two possible sources of error in the procedures used by Battelle: (1) the samples might have expanded or cracked while they were being oven-dried before the measurements were made causing an erroneously high porosity or (2) the samples might not have been totally saturated with  $\text{CCl}_4$  during the apparent density measurements which would give rise to erroneously low values. However, Battelle observed no change in the appearance or physical size of the samples upon drying. Because of the small sample size and the time allowed for saturating the sample, it seems unlikely that the apparent density would be measurably low; especially since Battelle's values were the highest. The values reported by Battelle are probably the most accurate.

The porosity and permeability determinations performed by Core Lab and Halliburton are both made using equipment and procedures developed by Core Lab. The porosity of core samples is measured in a helium porosimeter. After allowing two hours for the helium to permeate all of the open pores, the gauge reading appeared to be stable. However, considering the size of the sample and the number of days which were required to remove the contained natural gas from similar samples, it would seem that this method should yield erroneously low values. The values reported by Halliburton and Core Lab are both very low, but probably represent only the most accessible pore fraction.

The porosity values reported by the well logging companies result from two separate logging tools-- a neutron log and a density log. Density logs measure the concentration of electrons in a formation and yield fairly accurate bulk density values. Provided that the grain density of a rock and the density of any contained fluids are well defined, the porosity may also be found.

Neutron logging devices respond primarily to the concentration of hydrogen atoms in a volume rock. In general, these measurements are related to the amount of pore space filled with water or heavy hydrocarbons (gaseous hydrocarbons are not sufficiently dense to be detected). A better estimate of porosity may be obtained from cross-plots of the neutron and density log values for specific rock types. These cross-plot values are usually good approximations of the actual porosity.

A comparison of the porosimeter, gravimeter, and well log porosities shows that Battelle's values are very high (14.29% and 19.42%) for the two samples analyzed. Schlumberger's log readings are lower (.6% and 5.1%) for the corresponding depths, and the values from Core Lab and Halliburton for samples from the Lincoln County well are all extremely low (all less than 1.6%).

There was no correlation between the porosimetry readings and the well log porosities for the corresponding depths. However, it is difficult to ascertain whether the core and log depths are, in fact, the same. With only

two values from Battelle, it is not possible to develop a correlation but their bulk density values agree with the log values and are within the expected range. After reviewing the possibilities for error inherent in the methods, the technique used by Battelle appears to be the most sound.

#### Permeability Measurements

Juniata College, Battelle, and Core Lab have reported Devonian shale permeabilities. However, Juniata's research actually involves diffusion rates rather than true permeabilities. The Core Lab procedure measures the time required for a fluid of known viscosity to travel through a sample with a measured length and cross-section in response to a fixed pressure drive. On the other hand, Juniata's work suggests that the diffusion model should be more applicable to the capillary-sized passages in the shale.

#### Gas Content Measurements

The methods used to detect and measure gas contained in shale samples and the form in which the gas is held are of primary importance to reserve estimate calculations. Some gas may occur in interconnecting pores and will be escaping from the sample at ambient temperatures with time (free gas). Totally enclosed pore spaces may contain gas which is trapped until the pores are physically ruptured (trapped gas). Gas may also be sorbed onto mineral surfaces and/or organic matter within the shale. Any sorbed gas which occurs along the interconnected pore network may also be released with time. In addition, if the kerogen contained in rock samples is heated under proper conditions, it can break down to form gaseous and liquid hydrocarbons. Bacteria may also produce methane, ethylene, and propylene from certain organic materials.

Columbia has been concerned chiefly with measurements of the free gas content, although measurements of the trapped gas content and amount producible by kerogen breakdown are also being made. Only free and sorbed gas are presently considered as potentially available through hydraulic fracturing

since shale must be pulverized or heated to remove the trapped gas or degrade the kerogen.

Battelle determined the gas content in one shale sample from Belmont County, Ohio by thermogravimetric analyses at several temperatures. The gas volume was calculated from the weight loss of the sample (measured as a function of time), and the composition of the gases released was determined by mass spectrometry. As before, some gas was lost prior to canning, plus additional gas would have been lost when the core was removed from the can and the shale was cut to the proper size for analysis. In spite of these losses, they still reported a high gas content (30 cubic feet per cubic foot of shale). Subsequent investigations indicated that these values were far too high.

Well logs can also detect the presence of pore spaces occupied by gas by comparing neutron porosity and density logs, but the data cannot be used to derive quantitative estimates of the gas content.

#### Trapped Gas and Kerogen

Pyrolysis techniques are being used at Columbia Gas to study the changes in the trapped gas and kerogen content of shale samples. Colin Baker (1975) reports that as samples are heated, the gas pressure in closed pores increases until the enclosing mineral grains are forced apart. The released gas is detected as a distinct peak on a continuous trace from a total hydrocarbon detector. At higher temperatures, the gas produced from the breakdown of kerogen is similarly recorded. Several of our samples were also pulverized in sealed containers to measure the trapped gas content without applying heat.

Juniata College has also conducted some pyrolysis investigations. The results indicated an endothermic process occurring from 80°C to 113°C (interpreted as due to water desorption) and an exothermic process beginning at about 300°C which has been tied to the breakdown of kerogen. The gas produced from the kerogen breakdown contained large amounts of hydrogen, hydrogen sulfide, and carbon dioxide compared to the hydrocarbon content. From .05 to .41

millimoles of gas were produced per gram of shale and the calculated calorific value of the shale was from .66 to 36.1 calories per gram.

The validity of Columbia's gas content estimates rests on the assumption that the gas contained in the rock samples behaves as free gas in spite of whether the gas is actually free or adsorbed. This assumption appears to be justified from the results of previously mentioned experiments. However, the values may be subject to change when the open porosity has been verified by gravimetric methods. It is evident that for samples with such high gas contents that it is physically impossible for the total quantity to exist as a gas; the contents cannot be calculated by ideal gas laws.

There is also strong evidence to support the existence of an adsorbed gas phase within the shale, as shown by Paul Schettler of Juniata College. The adsorption model proposed by the Juniata research team is based on the work of Brunauer, Emmett and Teller. For all practical purposes, one should consider the shale "off-gas" and adsorbed gas to be one and the same.

APPENDIX II

APPENDIX II: GAS CONTENTS OF SHALE SAMPLES

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WELL #20402 LINCOLN COUNTY, WEST VIRGINIA

Point No.	Depth (Feet)	Off Gas	Trapped Gas	Kerogen
		Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale	Derived Gas Cu.Ft. Gas Cu.Ft. Shale
1	2660	.01		
2	2665	.00		
3	2670	.00		
4	2675	.00		
5	2680	.03		
6	2685	.03		
7	2690	.01		
8	2695	.03		
9	2700	.003	.09	1.12
10	2716	.00		
11	2721	.01	.15	1.53
12	2726	.00		
13	2731	.00		
14	2736	.00		
15	2741	.00	.21	2.84
16	2746	.02		
17	2751	.00		
18	2756	.09		
19	2766	.00		
20	3005	.11		
21	3010	.11		
22	3015	.08		
23	3020	.06		
24	3030	.15	1.07	1.25
25	3035	.11		
26	3040	.13		
27	3045	.20		
28	3055	.10		
29	3061	.21	.44	4.74
30	3066	.45		
31	3071	.33		
32	3076	.23		
33	3081	.12	.40	3.33
34	3086	.11		
35	3091	.27	.78	11.67
36	3096	.09		
37	3106	.17		
38	3111	.05		
39	3116	.50		
40	3298	.07	.47	1.23
41	3308	.63		
42	3313	.80		
43	3318	.74		
44	3323	.22		
45	3333	.10		
46	3338	.23		
47	3343	.13		
48	3348	.46		
49	3353	.00		

Point No.	Depth (Feet)	Off Gas	Trapped Gas	Kerogen Derived Gas
		Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale
50	3358	.26		
51	3363	.12		
52	3368	.21		
53	3373	1.16		
54	3383	.83		
55	3388	.22	2.48	9.49
56	3393	.09		
57	3398	1.22		
58	3403	.12		
59	3408	.26	3.35	49.27
60	3413	.00		
61	3418	.26	9.09	32.51
62	3423	.58		
63	3433	.08		
64	3438	.00	12.79	31.03
65	3443	.27		
66	3449	.80	8.07	25.40
67	3454	.15		
68	3459	.67	4.72	20.00
69	3464	.74		
70	3469	.82	7.05	21.65
71	3473	1.27		
72	3483	.29		
73	3488	.36		
74	3493	.35		
75	3498	.95	6.82	21.94
76	3503	.16		
77	3508	.38	3.50	25.43
78	3513	.08		
79	3518	.58		
80	3523	.74		
81	3532	.44	2.22	14.26
82	3537	.00		
83	3542	.00	3.35	20.97
84	3547	2.03		
85	3552	.45		
86	3557	.30		
87	3561	.21	1.68	5.94
88	3566	1.18		
89	3571	.08		
90	3576	.00		
91	3581	.46	4.72	26.43
92	3586	.73		
93	3901	.64		
94	3906	2.12	10.40	21.20
95	3911	2.59		
96	3917	.00		
97	3927	.39		
98	3937	1.91	.00	4.60
99	3943	1.32		
100	3948	.57		
101	3952	.04		
102	3957	.00		
103	3966	1.86		



WELL #20338 WISE COUNTY VIRGINIA

Point No.	Depth (Feet)	Off Gas	Trapped Gas	<sup>Yamner</sup> Derived Gas
		Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale
1	4871	3.64	.470	.820
2	4881	1.90	.460	.530
3	4890	2.95	.000	.000
4	4891	2.04	.210	.640
5	4901	3.22	1.190	3.260
6	4911	3.43	1.440	2.180
7	4921	2.34	.650	1.010
8	4931	1.82	.270	.950
9	4941	.17	.470	.560
10	4951	.69	.020	.100
11	4961	.00	.050	.110
12	4971	.00	.020	.120
13	4981	.15	.070	.060
14	5211	.61	.520	.280
15	5221	.45	.040	.130
16	5231	.39	.110	.490
17	5241	.61	.280	.390
18	5251	.97	.100	.120
19	5261	.13	.050	.030
20	5271	2.16		
21	5281	1.47	.260	.930
22	5291	.03	.550	1.550
23	5302	2.93	.180	.090
24	5311	.26	.410	1.110
25	5321	1.86	.620	.810
26	5331	.66	.002	.050
27	5341	2.72	.510	1.170
28	5350	2.34	.070	.470
29	5361	.02	.080	.120
30	5363	4.49		
31	5366	2.76		
32	5371	3.24	1.060	.790
33	5380	1.26	.120	.440
34	5401	4.20	.930	1.270
35	5411	.70	.230	.600
36	5421	3.17	.750	1.820
37	5431	2.20	.060	.050
38	5441	3.87		
39	5445	2.93		
40	5451	4.09	.750	.910
41	5461	1.78	.530	2.430
42	5471	3.10	.320	.800
43	5688	.11	.040	.090

WELL #20336 MARTIN COUNTY, KENTUCKY

Point No.	Depth (Feet)	Off Gas	Trapped Gas	Kerogen
		Cu. Ft. Gas Cu. Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale	Derived Gas Cu.Ft. Gas Cu.Ft. Shale
1	2442	.64		
2	2452	.06	4.424	5.547
3	2462	.11	1.137	.284
4	2472	.00	2.097	8.059
5	2482	1.78	2.157	7.365
6	2490	.00	3.063	7.775
7	2493	.71	2.768	10.400
8	2503	.26	2.299	8.252
9	2513	.02	.227	.810
10	2523	.00	.114	.213
11	2533	.00	.000	.099
12	2543	.19	.037	.043
13	2549	.10	.028	.205
14	2553	.00	.023	.085
15	2564	.00	.026	.109
16	2574	.00	.234	.386
17	2584	.00	.043	.097
18	2587	.32	1.279	3.538
19	2594	.00	.016	.017
20	2604	.00	.158	.473
21	2624	.02	.112	.261
22	2644	.00	.061	.033
23	2645	.08	.081	.274
24	2654	.00	.189	.426
25	2664	.00	.099	.321
26	2674	.00	3.717	13.515
27	2684	.00	2.813	8.866
28	2694	.00	3.740	10.162
29	2704	.00	5.044	15.583
30	2716	.00	3.373	7.490
31	2725	.03	.301	.367
32	2735	.00	1.989	3.279
33	2745	.00	.922	2.984
34	2755	.05	1.515	4.217
35	2762	.00	.999	3.191
36	2765	.00	2.262	8.445
37	2775	.00	2.521	8.297
38	2785	.00	3.728	8.763
39	2795	.00	.313	.578
40	2805	.04	.250	.261
41	2815	.02	.386	.277
42	2820	.06	.199	.125
43	2825	.00	1.918	5.479
44	2835	.00	.159	.253
45	2845	.00	1.831	5.069
46	2855	.68	3.308	6.786
47	2865	.00	.469	.483

Point No.	Depth (Feet)	Off Gas	Trapped Gas	Kerogen
		Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale	Derived Gas Cu.Ft. Gas Cu.Ft. Shale
48	2875	.09	2.606	5.001
49	2879	.06	1.091	2.141
50	2885	.00	2.404	7.070
51	2895	.36	.335	.148
52	2905	.09	2.373	5.456
53	2915	.21	3.637	10.105
54	2925	.00	4.015	11.707
55	2930	.14	.069	.029
56	2935	.00	.591	1.157
57	2940	.74	3.632	7.945
58	2947	.27	3.779	11.776
59	2957	.28	2.427	6.706
60	2967	2.02	3.666	11.889
61	2977	.87	2.827	10.150
62	2987	.05	3.168	9.968
63	2997	1.23	3.518	13.071
64	2998	1.53	3.714	17.175
65	3008	.32	.904	3.893
66	3018	.00	3.151	11.128
67	3028	.00	.435	.340
68	3034	.00	1.068	1.571
69	3038	1.02	7.127	17.823
70	3048	.00	5.388	16.174
71	3058	.00	3.032	15.049
72	3068	.00	1.938	14.958
73	3078	.00	3.694	13.128
74	3088	2.05	2.751	14.958
75	3098	1.97	4.319	15.004
76	3105	.05	.696	2.657
77	3107	.00	4.225	14.799
78	3117	.04	7.331	22.585
79	3127	.00	.189	.166
80	3137	.00	.308	.767
81	3147	.13	.088	.646
82	3157	.11	.195	.000
83	3163	.00	.037	.016
84	3168	.07	.412	1.165
85	3178	.25	.074	.563
86	3188	.00	.021	.115
87	3198	.00	2.586	7.388
88	3209	.00	.007	.000
89	3218	.82	.936	1.077
90	3220	.00	.114	.017
91	3227	.00	.141	.695
92	3237	.04	.027	.833
93	3247	.00	.028	.024
94	3257	.08	.027	.037
95	3267	.12	.125	.632
96	3277	.00	.092	.000
97	3280	.16	.028	.017

Point No.	Depth (Feet)	Off Gas	Trapped Gas	Kerogen
		Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale	Derived Gas Cu.Ft. Gas Cu.Ft. Shale
98	3287	.30	4.910	13.230
99	3297	.11	.065	.014
100	3307	.04	.081	.134
101	3317	.34	.142	.097
102	3317	.14	.000	.000
103	3328	.00	.149	.563
104	3338	.06	4.274	13.003
105	3348	.05	.028	.156
106	3358	.80	1.770	15.174
107	3368	.20	4.098	17.936
108	3378	.64	5.700	21.369
109	3379	.51	.171	1.745
110	3388	.75	3.154	7.873
111	3398	.00	1.344	3.513
112	3408	.00	.210	.371

WELL #20403 LINCOLN COUNTY, WEST VIRGINIA

Point No.	Depth (Feet)	Off Gas	Trapped Gas	Kerogen
		Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale	Derived Gas Cu.Ft. Gas Cu.Ft. Shale
1	2735	.00	.074	.551
2	2740	.01	.050	.594
3	2745	.00	.124	1.729
4	2750	.03		
5	2755	.01		
6	2760	.00		
7	2765	.00		
8	2770	.00		
9	2775	.03	.034	.446
10	2782	.03		
11	2787	.04	.077	1.165
12	2792	.00	.355	.000
13	2797	.00	.105	.906
14	2802	.00	.071	.833
15	2807	.04	.073	.997
16	2812	.03		
17	2817	.00	.055	.727
18	2827	.05	.311	2.565
19	2832	.05	.148	10.593
20	2837	.00	.125	3.626
21	2842	.05	.109	.698
22	2847	.08	.091	2.305
23	2852	.02	.054	.438
24	2857	.14	.256	.000
25	2862	.00	.219	2.475
26	2872	.00	.190	2.319
27	2877	.00	.119	1.554
28	2882	.03		
29	2887	.11	1.097	.199
30	2892	.00	1.279	.102
31	2897	.07		
32	2912	.14	.870	.085
33	2917	.29		
34	2918	.00		
35	2922	.00	1.284	8.627
36	2927	.00		
37	2933	.03	.398	1.489
38	2938	.03		
39	2943	.00	.651	2.907
40	2947	.00		
41	2952	.01		
42	2959	.05		
43	2964	.05		
44	2969	.14		
45	2974	.11	2.694	10.336
46	2979	.05		
47	2984	.17		
48	2989	.05		

Point No.	Depth (Feet)	Off Gas	Trapped Gas	Kerogen
		Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale	Derived Gas Cu.Ft. Gas Cu.Ft. Shale
49	2994	.06	.546	8.082
50	2999	.01		
51	3004	.08	6.695	29.837
52	3009	.00		
53	3014	.17	.447	3.876
54	3019	.13		
55	3024	.11	.995	7.184
56	3025	.00		
57	3029	.15		
58	3034	.14	1.807	11.821
59	3035	.09		
60	3039	.12		
61	3044	.08	.648	2.251
62	3049	.10		
63	3054	.08	.466	4.643
64	3059	.09		
65	3064	.09	.989	6.700
66	3081	.00		
67	3085	.22		
68	3090	.00		
69	3095	.16	3.183	12.651
70	3100	.20		
71	3106	.48		
72	3111	.22		
73	3116	.15	.602	6.161
74	3121	.11		
75	3126	.22	1.080	8.127
76	3131	.19		
77	3135	.21	2.876	13.128
78	3140	.09		
79	3145	.06	.887	5.104
80	3150	.00		
81	3155	.14	.557	7.275
82	3160	.03		
83	3165	.06	2.205	9.434
84	3170	.16		
85	3175	.11	.932	5.251
86	3180	.09		
87	3185	.24	1.364	9.866
88	3190	.27		
89	3198	.14	.318	2.216
90	3203	.22		
91	3208	.04	.284	1.898
92	3213	.00		
93	3218	.15	.227	2.228
94	3223	.09		
95	3228	.10	.369	1.353
96	3233	.00		
97	3238	.05	.281	1.824
98	3243	.04		
99	3248	.16	1.853	10.139

Point No.	Depth (Feet)	Off Gas		Trapped Gas		Kerogen Derived Gas	
		Cu.Ft. Gas	Cu.Ft. Shale	Cu.Ft. Gas	Cu.Ft. Shale	Cu.Ft. Gas	Cu.Ft. Shale
100	3254	.00					
101	3263	.12		.216		1.057	
102	3268	.00					
103	3273	.00		.727		3.651	
104	3276	.06					
105	3280	.29		.261		2.637	
106	3285	.08					
107	3290	.00		.239		1.654	
108	3295	.00					
109	3300	.10		.344		1.284	
110	3305	.04					
111	3310	.07		.256		1.165	
112	3316	.15					
113	3321	.00		1.444		16.368	
114	3336	.14					
115	3341	.19		3.762		11.673	
116	3346	.29					
117	3351	.12		.943		2.194	
118	3356	.11					
119	3361	.00		.494		1.540	
120	3366	.41					
121	3371	.20					
122	3376	.32					
123	3381	.18		.318		1.239	
124	3386	.10					
125	3391	.13		.222		.779	
126	3396	.11					
127	3401	.00		.318		.585	
128	3406	.12					
129	3411	.14		5.297		14.992	
130	3416	.11					
131	3421	.15		5.592		13.356	
132	3432	.34		7.922		21.835	
133	3437	.66					
134	3442	.53		.421		3.171	
135	3447	.78					
136	3452	.37		6.706		14.231	
137	3457	.49					
138	3462	.33		2.501		8.411	
139	3467	.21					
140	3472	.14		.864		2.035	
141	3477	.45					
142	3482	.31		1.193		7.559	
143	3487	1.03					
144	3492	.22		1.444		13.981	
145	3498	.08					
146	3503	.20					
147	3508	.38					
148	3513	.25		7.024		14.310	
149	3518	.20					
150	3523	.89		3.626		17.720	
151	3528	.18					
152	3533	.31		8.059		19.425	

Point No.	Depth (Feet)	Off Gas	Trapped Gas	Kerogen
		Cu.Ft. Gas Cu.Ft. Shale	Cu.Ft. Gas Cu.Ft. Shale	Derived Gas Cu.Ft. Gas Cu.Ft. Shale
153	3538	.65		
154	3543	.04		
155	3548	.17		
156	3553	.26	5.820	13.787
157	3558	1.16		
158	3563	.37	7.286	19.630
159	3568	.62		
160	3573	.25	7.024	21.494
161	3578	.35		
162	3583	.32		
163	3588	.250		
164	3598	.81		
165	3606	.21	2.467	12.980
166	3611	.19		
167	3616	.36	3.865	18.016
168	3621	.23		
169	3626	.12	3.433	12.333
170	3631	.19		
171	3636	.27	10.159	26.597
172	3641	.50	.008	.023
173	3646	.24		
174	3651	.12		
175	3656	.10		
176	3661	.62		
177	3666	.00		
178	3671	.00		
179	3676	.06		
180	3681	.07		
181	3686	.14		
182	3691	.02		
183	3696	.26		
184	3701	.21		
185	3706	.34		
186	3711	.00		
187	3716	.01		
188	3721	.03	.047	.438
189	3726	.004		
190	3732	.00		
191	3737	.18		
192	3743	.04		
193	3748	.40		
194	3753	.03		
195	3759	1.42		
196	3764	.06		
197	3768	1.32		
198	3773	1.02		
199	3777	.19		
200	3782	1.23		
201	3787	.05	.182	.375
202	3792	.09		
203	3797	.01	.003	.123



Point No.	Depth (Feet)	Off Gas Cu.Ft. Gas Cu.Ft. Shale	Trapped Gas Cu.Ft. Gas Cu.Ft. Shale	Kerogen Derived Gas Cu.Ft. Gas Cu.Ft. Shale
204	3802	.00		
205	3807	.60	.119	.905
206	3812	.02		
207	3818	.02	.026	.055
208	3823	.11		
209	3829	.00	.025	.388
210	3834	.01		
211	3839	.00	.202	.000
212	3844	.48		
213	3849	.08	.010	.065
214	3854	.08		
215	3859	1.69	1.080	2.961
216	3864	.07		
217	3869	.08	.051	.079
218	3874	.06		
219	3879	.53		
220	3884	.07		
221	3888	.06	.094	1.502
222	3893	.40		
223	3898	.15	.187	.764
224	3903	1.47		
225	3908	.19	1.176	1.489
226	3913	.67		
227	3918	.17	2.569	1.563
228	3923	.65		
229	3928	.46	2.103	4.205
230	3934	.33		
231	3939	.16	.824	.960
232	3944	.24		
233	3950	.28	1.040	2.756
234	3955	.80		
235	3959	.78	4.137	8.525
236	3964	1.02		
237	3969	.81	3.546	6.081
238	3974	2.21		
239	3979	1.35	8.400	10.650
240	3984	1.13		
241	3998	.57	6.615	6.956
242	4003	.00		
243	4010	1.35	4.091	7.002
244	4015	.55		
245	4020	.96	2.296	4.467
246	4025	.03		
247	4025	.25		
248	4030	2.16	1.767	4.242
249	4030	2.09		
250	4035	1.23		
251	4035	1.22		
252	4040	1.05	.821	2.495
253	4049	2.28		
254	4051	.59		

APPENDIX III

APPENDIX III: CORE DESCRIPTIONS

Well No. 20403

Core #1 (2720' - 2779'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2720'	2779'6"	Medium gray-green laminated silty shale.

<u>From</u>	<u>To</u>	<u>Fracture</u>
2722'6"	2722'7"	Vertical
2723'11"	2724'2"	Vertical
2728'7½"	2728'9 3/4"	Vertical
2735'0"	2735'3"	Vertical
2742'4"	2742'10"	Vertical
2744'2"	2744'4"	Vertical
2748'1½"	2748'5"	Vertical
2751'1½"	2751'3½"	Vertical
2751'7½"	2751'9½"	Vertical
2764'6"	2764'8½"	Vertical
2778'½"	2778'2½"	Vertical

Distinct Black Kerogen Layers

2769'

Comments: Mud logger reported no indication of gas.

## Core #2 (2779'6" - 2838')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2779'6"	2795'6"	Medium dark gray laminated silty shale
2795'6"	2796'6"	Dark gray shale
2796'6"	2804'6"	Laminated medium gray silty shale
2804'6"	2805'6"	Medium green-gray shale
2805'6"	2809'6"	Medium dark gray silty shale
2809'6"	2821'6"	Medium to dark gray laminated silty shale
2821'6"	2823'6"	Medium gray shale
2823'6"	2831'6"	Medium and dark gray laminated silty shale
2831'6"	2832'6"	Medium gray shale
2832'6"	2838'	Medium dark gray silty shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2783'11"	2784'11"	Vertical
2792'9"	2793'7"	Vertical
2794'7"	2795'3"	Vertical, partially open
2828'6"	2829'6"	Vertical, partially open
2829'9"	2831'2"	Vertical

Comments: Mud logger reported no indications of gas.

Core #3 (2838' - 2896'4")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2838'	2853'	Medium to dark gray laminated silty shale but less silty than cores 1 and 2
2853'	2855'	Dark gray shale
2855'	2866'	Medium dark gray laminated silty shale
2866'	2867'	Dark gray shale
2867'	2874'	Dark gray laminated silty shale
2874'	2876'3"	Dark gray shale with two silty layers ¼" thick
2876'3"	2881'8"	Dark gray laminated silty shale
2881'8"	2894'	Dark gray shale with light and dark beds grading downward into very dark gray shale
2894'	2896'4"	Very dark gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2892'9"	2894'11"	Vertical, open, with gas bleeding along fracture surface

Comments: Mud logger reported no indications of gas. Core crumbled in places.

## Core #4 (2896'4" - 2955'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2896'4"	2904'4"	Very dark gray shale with several silty laminae
2904'4"	2908'4"	Dark gray shale with abundant dispersed silty layers
2908'4"	2910'4"	Very dark gray shale
2910'4"	2920'11"	Dark gray shale with dispersed lighter silty layers $\frac{1}{4}$ " to 1"
2920'11"	2926'4"	Very dark gray shale with one 2" silty layer
2926'4"	2931'4"	Dark gray shale with dispersed silty layers
2931'4"	2932'4"	Very dark gray shale
2932'4"	2937'4"	Dark gray shale with dispersed silty layers
2937'4"	2938'4"	Very dark gray shale
2938'4"	2945'7 $\frac{1}{2}$ "	Dark gray shale with dispersed silty layers
2945'7 $\frac{1}{2}$ "	2948'4"	Very dark gray shale
2948'4"	2955'6"	Alternating bands of medium gray and very dark gray shale with occasional silty layers

Comments: No indications of gas reported by mud logger. Core crumbled in places.

Core #5 (2955'6" - 3014')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2955'6"	2962'6"	Very dark gray shale
2962'6"	2964'6"	Medium dark gray shale with dispersed silt and carbonate stringers
2964'6"	2975'6"	Very dark gray shale with four silty layers
2975'6"	2981'6"	Light to medium gray shale with several 1" silt-carbonate layers
2981'6"	2987'6"	Very dark gray shale
2987'6"	2999'6"	Very dark gray shale with dispersed ¼" silt-carbonate stringers
2999'6"	3014'	Very dark gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2955'6"	2959'10"	Vertical and open
2968'8"	2972'9½"	Vertical and open
3009'8"	3013'6"	Vertical and open

Distinct Black Kerogen Layers

2981'  
2995'  
2996'9"  
3000'  
3005'

Comments: Mud logger reported no indication of gas. Very dark gray shale is brown when reduced to powder.

## Core #6 (3014'-3073')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3014'	3017'	Dark gray shale
3017'	3029'6"	Medium gray shale with siltier layers up to 2", pyrite layer along bedding plane at 3018'
3029'6"	3039'	Medium to dark gray shale
3039'	3053'	Medium to dark gray silty shale
3053'	3073'	Medium to dark gray shale with occasional silty shale horizons

<u>From</u>	<u>To</u>	<u>Fractures</u>
3039'6"	3046'6"	Vertical, mineralized
3049'3"	3053'6"	Vertical
3049'3"	3063'6"	Vertical, mineralized
3053'6"	3033'6"	Vertical
3065'3½"	3073'	Vertical

Distinct Black Kerogen Layers

3030'

Comments: Mud logger detected "trip gas" after coring this interval.



Core #7 (3073'-3132'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3073'	3132'6"	Medium to dark gray shale with occasional darker bands - only rare and thin silty layers.

<u>From</u>	<u>To</u>	<u>Fractures</u>
3073'	3084'	Vertical, mineralized
3079'	3119'1"	Vertical
3119'1"	3126'6"	Vertical
3126'6"	3130'6"	Vertical

## Core #8 (3132'6"-3192')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3132'6"	3140'4"	Medium gray shale with 1" dark gray bands
3140'4"	3140'5-1/2"	Light gray, very silty shale to siltstone
3140'5-1/2"	3144'7"	Dark gray shale
3144'7"	3144'8-1/2"	Light gray very silty shale to siltstone
3144'8-1/2"	3150'1-1/2"	Dark to medium gray shale
3150'1-1/2"	3150'4"	Light gray, very silty shale to siltstone
3150'4"	3155'8"	Dark and medium dark gray shale
3155'8"	3155'11"	Light gray, very silty shale to siltstone
3155'11"	3165'1"	Dark to medium dark gray shale with numerous light gray siltstone laminae.
3165'1"	3165'5"	Light gray crossbedded argillaceous siltstone with dark gray shale bands
3165'5"	3183'5"	Dark to medium gray shale with numerous light gray siltstone bands up to 1" thick
3183'5"	3183'6"	Very light gray siltstone
3183'6"	3192'	Very dark gray to black shale (brown when reduced to powder)

<u>From</u>	<u>To</u>	<u>Fractures</u>
3140'	3140'6"	Vertical and open
3141'	3142'	Vertical and open
3146'	3146'10"	Vertical and open
3148'	3150'	Vertical and open
3151'8"	3152'7"	Vertical and open
3154'	3156'7"	Vertical and open
3160'8"	3161'4"	Vertical and open
3164'6"	3165'	Vertical and open
3167'8"	3169'	Vertical and open
3169'8"	3170'8"	Vertical and open
3171'	3172'	Vertical and open
3173'3"	3174'	Vertical and open
3184'	3186'4"	Vertical and open
3186'6"	3188'10"	Vertical and open
3189'	3192'	Vertical and open

Core #9 (3192'-3251'1")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3192'	3197'9"	Dark gray to black shale with a few medium gray 1" thick layers
3197'9"	3200'2"	Medium gray shale with several 1" dark and very dark gray layers
3200'2"	3200'4-1/2"	Light gray argillaceous siltstone
3200'4-1/2"	3220	Medium and dark gray laminated shale with dark gray shale in 1/4" bands, three silt layers to 2" thick
3220'	3221'9"	Very dark gray to black shale
3221'9"	3227'	Medium gray and very dark gray banded shale
3227'	3230'3"	Medium gray shale with numerous very dark gray bands to 2" thick
3230'3"	3230'5-1/2"	Light gray argillaceous siltstone
3230'5-1/2"	3230'10-1/2"	Very dark gray to black shale
3230'10-1/2"	3246'	Medium and dark gray banded shale
3246'	3246'9"	Light gray argillaceous siltstone
3246'9"	3251'1"	Medium to very dark gray banded shale, very dark gray bands to 4" thick

<u>From</u>	<u>To</u>	<u>Fractures</u>
3192'2"	3193'3"	Vertical and open
3193'10"	3194'6"	Vertical and open
3194'6"	3197'	Vertical and open
3197'2"	3198'8"	Vertical and open
3200'7"	3202'6"	Vertical and open
3202'7"	3203'4"	Vertical and open
3208'4"	3209'9"	Vertical and open
3210'7"	3211'4"	Vertical and open
3212'	3213'	Vertical and open
3213'	3213'7"	Vertical and open
3215'10"	3216'10"	Vertical and open
3217'3"	3218'3"	Vertical and open
3218'4"	3219'3"	Vertical and open
3222'3"	3222'9"	Vertical and open
3225'6"	3227'2"	Vertical and open
3228'8"	3230'3"	Vertical and open
3230'7"	3239'1"	Vertical and open
3239'3"	3240'6"	Vertical and open
3241'3"	3244'8"	Vertical and open
3244'8"	3246'	Vertical and open
3248'	3251'1"	Vertical and open

## Core #10 (3251'1"-3310'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3251'1"	3260'6"	Medium to dark gray shale with very dark gray to black bands up to 1-1/2" thick
3260'6"	3260'11"	Light gray argillaceous siltstone
3260'11"	3261'8"	Laminated medium gray shale and argillaceous siltstone
3261'8"	3310'3"	Very dark gray to black shale
3310'3"	3310'6"	Light gray siltstone

<u>From</u>	<u>To</u>	<u>Fractures</u>
3255'	3258'10"	Vertical and open
3259'4"	3260'6"	Vertical and open
3261'	3261'6"	Vertical and open
3262'	3262'10"	Vertical and open
3264'10"	3271'4"	Vertical and open
3271'7"	3274'6"	Vertical and open
3275'2"	3280'9"	Vertical and open
3280'11"	3283'6"	Vertical and open
3283'6"	3289'4"	Vertical and open
3289'7"	3291'6"	Vertical and open
3291'10"	3293'9"	Vertical and open
3293'9"	3297'	Vertical and open
3297'4"	3300'4"	Vertical and open
3300'4"	3301'8"	Vertical and open
3301'10"	3307'7"	Vertical and open
3309'3"	3310'5"	Vertical and open

Core #11 (3310'6" - 3368'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3310'6"	3314'2"	Medium gray shale with dark gray bands to 2" thick, severely broken
3314'2"	3323'10"	Dark and very dark gray shale with four 5" black bands
3323'10"	3336'4"	Medium gray shale with dark gray bands
3336'4"	3344'5"	Very dark gray to black shale (with brown streak) and medium gray bands
3344'5"	3344'7"	Light gray silty shale
3344'7"	3346'5"	Very dark gray to black shale (with brown streak)
3346'5"	3352'6"	Medium gray shale with very dark gray to black shale bands up to 8" thick
3352'6"	3363'6"	Medium gray shale with very dark and black bands to 1" thick, and silty layers to 1" in thickness
3363'6"	3366'	Dark gray shale with bands of medium and very dark gray shale, several light gray silty layers to 1" thick
3366'	3368'6"	Very dark gray to black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3347'8"	3348'10"	Vertical and open
3357'2"	3358'3"	Vertical and open
3310'6"	3314'2"	Vertical and open
3314'2"	3315'1"	Vertical and open
3314'1"	3317'8"	Vertical and open
3317'8"	3319'6"	Vertical and open
3319'6"	3321'5"	Vertical and open
3321'6"	3323'11"	Vertical and open
3325'5"	3326'6"	Vertical and open
3326'6"	3331'	Vertical and open
3337'6"	3338'9"	Vertical and open
3338'9"	3339'10"	Vertical and open
3340'4"	3342'4"	Vertical and open
3355'8"	3356'9"	Vertical and open
3358'2"	3359'4"	Vertical and open

## Core #12 (3368'6" - 3428')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3368'6"	3377'8"	Black shale (brown streak)
3377'8"	3377'9"	Light gray silty shale
3377'9"	3378'6"	Black shale
3378'6"	3381'6"	Medium gray shale with black bands, several with silty layers
3381'6"	3382'1"	Black shale
3382'1"	3391'4"	Medium gray shale with black bands several silty layers
3391'4"	3391'7"	Medium gray siltstones
3391'7"	3409'10"	Medium gray shale with black bands to 2" thick, and several light gray siltstones to 1"
3409'10"	3424'1"	Black shale with calcite veinlets
3424'1"	3428'	Very dark gray and black banded shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3368'6"	3370'10"	Vertical and open
3370'11"	3372'1"	Vertical and open
3371'11"	3373'9"	Vertical and open
3374'11"	3376'8"	Vertical and open
3379'5"	3381'4"	Vertical and open
3388'	3395'5"	Vertical and open
3398'2"	3399'3"	Vertical and open
3411'2"	3412'6"	Vertical and open
3412'6"	3415'4"	Vertical and open
3416'4"	3419'	Vertical and open
3419'	3424'	Vertical and open
3424'3"	3425'3"	Vertical and open

Distinct Black Kerogen Layer

3422'5-1/2"

Core #13 (3428' - 3487')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3428'	3448'9"	Black shale
3448'9"	3449'6"	Medium gray shale
3449'6"	3449'9"	Black shale
3449'9"	3450'9 1/2"	Medium gray shale
3450'9 1/2"	3451'5"	Black shale
3451'5"	3451'9"	Medium gray shale
3451'9"	3460'5"	Black shale
3460'5"	3462'5"	Medium gray shale with two 2" black bands
3462'5"	3468'5"	Medium gray shale with six black bands to 4", six silty layers to 2" and pyrite blebs and layers
3468'5"	3471'5"	Black shale
3471'5"	3474'5"	Medium gray and black shale bands
3474'5"	3479'1"	Black shale
3479'1"	3479'2"	Medium gray shale
3479'2"	3479'3"	Silty zone
3479'3"	3481'	Black shale
3481'	3483'8"	Dark gray and black shale
3483'8"	3484'10"	Medium gray shale
3484'10"	3485'1"	Blue gray siltstone
3485'1"	3485'4"	Medium gray shale
3485'4"	3486'3"	Dark gray shale
3486'3"	3487'	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3428'9"	3431'	Vertical and open
3430'9"	3435'5"	Vertical and open
3435'5"	3436'8"	Vertical and open
3437'4"	3438'6"	Vertical and open
3438'4"	3441'8"	Vertical and open
3441'8"	3444'3"	Vertical and open
3444'1"	3449'	Vertical and open
3452'2"	3460'6"	Vertical and open
3469'4"	3472'2"	Vertical and open
3461'8"	3467'	Vertical and open
3467'5"	3469'	Vertical and open

Distinct Black Kerogen Layers

3428'9  
3460'1"  
3471'  
3478'

## Core #14 (3487' - 3546'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3487'	3487'9"	Medium gray shale with 1" silty layer
3487'9"	3493'5"	Black shale
3493'5"	3495'8"	Medium green-gray shale with two 1" black bands
3495'8"	3496'3"	Numerous pyrite blebs in black shale, one 1/4" pyrite stringer
3496'3"	3502'	Medium green gray shale, pyrite blebs, and one 1/2" and one 1-1/2" silty layer
3502'	3502'8"	Black shale
3502'8"	3503'11"	Medium green-gray shale with a 1" black shale band
3503'11"	3504'2"	Medium gray argillaceous siltstone
3504'2"	3504'10"	Medium gray silty shale
3504'10"	3505'10"	Black shale
3505'10"	3506'4"	Contorted black shale and siltstone
3506'4"	3508'5"	Hard, medium to light gray-green siltstone
3508'5"	3509'4"	Medium and black banded shale
3509'4"	3510'2"	Black shale
3510'2"	3510'6"	Medium gray and black shale (contorted bedding)
3510'6"	3510'10"	Black shale
3510'10"	3511'	Medium gray and black contorted shale
3511'	3518'1"	Black shale with a 1" pyrite band
3518'1"	3520'2"	Gray-green shale with two 3" black bands
3520'2"	3520'9"	Black shale
3520'9"	3521'9"	Green-gray shale with light gray silty layers
3521'9"	3523'3"	Black shale with 5 dark gray-green bands to 1" thick
3523'3"	3524'11"	Black shale
3524'11"	3525'3"	Medium gray-green shale
3525'3"	3529'3"	Black shale
3529'3"	3529'5"	Gray-green shale
3529'5"	3532'1"	Black shale
3532'1"	3532'4"	Gray-green shale
3532'4"	3533'9"	Black shale
3533'9"	3534'11"	Laminated gray-green and black shale
3534'11"	3541'	Black shale
3541'	3541'2"	Gray-green shale
3541'2"	3542'2"	Black shale
3542'2"	3544'8"	Medium gray green shale with 3 black layers 1/8" to 1" thick
3544'8"	3545'1"	Black shale with pyrite bands and blebs
3545'1"	3546'6"	Medium gray-green shale with one 2" black band



Core #14 (Cont'd)

<u>From</u>	<u>To</u>	<u>Fractures</u>
3489'	3492'1"	Vertical and open
3492'1"	3494'	Vertical and open
3494'10"	3501'6"	Vertical and open
3502'2"	3503'5"	Vertical and open
3508'6"	3519'1"	Vertical and open
3521'9"	3527'4"	Vertical and open
3527'1"	3531'6"	Vertical and open
3531'6"	3538'6"	Vertical and open
3538'6"	3541'7"	Vertical and open
3544'1"	3546'5"	Vertical and open

Several Distinct Black Kerogen Layers

Complete distinct layers and partial layers were observed at several depths.

Core #15 (3546'6" - 3606')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3546'6"	3553'	Medium to dark gray shale with parallel black layers (1/4" to 3")
3553'	3564'5"	Black shale with three 1" medium gray bands
3564'5"	3569'5"	Medium gray shale with four black bands (1/4" to 3")
3569'5"	3570'6"	Black shale
3570'6"	3570'9"	Medium gray shale
3570'9"	3570'11"	Medium gray siltstone
3570'11"	3571'5"	Black shale with pyrite
3571'5"	3571'8"	Medium gray shale
3571'8"	3580'10"	Black shale
3580'10"	3582'10"	Medium dark gray and black shale band 4-7" thick
3582'10"	3590'9"	Black shale
3590'9"	3593'4"	Black shale with five medium dark gray bands 3-5" thick
3593'4"	3595'6"	Black shale
3595'6"	3595'10"	Medium gray shale
3595'10"	3596'3"	Black shale
3596'3"	3596'11"	Medium gray shale
3596'11"	3598'8"	Black shale
3598'8"	3598'10"	Contorted medium gray shale
3598'10"	3599'4"	Black shale
3599'4"	3599'6"	Medium gray shale
3599'6"	3606'	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3548'	3551'8"	Vertical and open
3555'5"	3557'6"	Vertical and open
3557'8"	3562'7"	Vertical and open
3572'	3580'	Vertical and open
3582'11"	3585'3"	Vertical and open
3585'2"	3588'	Vertical and open
3589'	3589'4"	Vertical and open
3589'4"	3591'10"	Vertical and open
3596'8"	3598'	Vertical and open

Core #16. (3606'-3665')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3606'	3612'3"	Black shale with contorted dark gray laminae
3612'3"	3613'	Medium dark gray shale
3613'	3613'5"	Black shale
3613'5"	3614'2"	Medium gray shale
3614'2"	3619'7"	Black shale
3619'7"	3620'3"	Medium gray shale
3620'3"	3620'6"	Black shale
3620'6"	3620'9"	Medium gray shale with silty layers
3620'9"	3621'3"	Black shale
3621'3"	3621'9"	Medium gray shale with black laminae
3621'9"	3623'5"	Black shale
3623'5"	3623'8"	Dark gray shale
3623'8"	3624'	Black shale
3624'	3624'9"	Medium gray shale
3624'9"	3625'5"	Black shale
3625'5"	3625'9"	Medium gray shale with black laminae
3625'9"	3628'3"	Black shale
3628'3"	3628'9"	Medium dark shale with black laminae
3628'9"	3629'4"	Black shale
3629'4"	3630'	Laminated dark gray and black shale
3630'	3632'5"	Black shale
3632'5"	3632'10"	Very dark shale with a 1/4" siltstone layer at base
3632'10"	3633'4"	Black shale
3633'4"	3633'9"	Very dark shale
3633'9"	3644'1"	Black shale with several very dark layers
3644'1"	3644'6-1/2"	Very dark gray shale
3644'6-1/2"	3644'8"	Medium gray argillaceous siltstone
3644'8"	3647'3"	Black shale
3647'3"	3648'	Black and very dark gray shale, one large pyrite bleb
3648'	3650'4"	Black shale
3650'4"	3650'11"	Black and dark gray shale
3650'11"	3651'2"	Black shale
3651'2"	3651'4"	Very dark gray shale
3651'4"	3651'7"	Black shale
3651'7"	3651'10"	Very dark gray shale
3651'10"	3652'	Black shale
3652'	3652'7"	Dark gray shale
3652'7"	3653'7"	Black shale
3653'7"	3653'9"	Very dark gray shale
3653'9"	3654'	Black shale
3654'	3654'4"	Very dark gray shale
3654'4"	3654'10"	Black shale
3654'10"	3655'	Very dark gray shale
3655'	3659'11"	Black shale
3659'11"	3661'7"	Medium gray shale with black laminae
3661'7"	3661'10"	Black shale
3661'10"	3662'2"	Medium dark gray shale

## Core #16 (Cont'd)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3662'2"	3663'3"	Black shale with a 1/8" pyrite layer
3663'3"	3663'8"	Dark gray shale with black laminae
3663'8"	3663'11"	Black shale with 1/8" pyrite stringer at base
3663'11"	3664'9"	Medium gray shale
3664'9"	3665'	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3606'	3606'5"	Vertical, open
3608'2"	3609'11"	Vertical, open
3611'	3612'3"	Vertical, open
3615'3"	3620'7"	Vertical, open
3621'9"	3622'11"	Vertical, open
3630'2"	3631'7"	Vertical, open
3631'7"	3632'9"	Vertical, open
3632'9"	3635'2"	Vertical, open
3635'10"	3637'10"	Vertical, open
3638'3"	3639'10"	Vertical, open
3644'2"	3646'7"	Vertical, open
3648'	3650'8"	Vertical, open
3652'8"	3656'3"	Vertical, open
3656'3"	3659'3"	Vertical, open

Distinct Black Kerogen Layers:

3617'4"  
 3621'  
 3636'  
 3643'7"  
 3657'8"

Comment: Mud logger reported increase in gas show during Core #16.

Core #17 (3665'-3724'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3664'	3665'	Black shale
3665'	3677'2"	Blue gray shale, abundant pyrite, numerous darker bands to 1" thick, a 1" silty layer at 3675'5".
3677'2"	3677'5"	Dark gray shale
3677'5"	3683'5"	Laminated dark gray and blue-gray shale
3683'5"	3684'5"	Blue-gray shale
3684'5"	3688'1"	Blue-gray and dark gray laminated shale
3688'1"	3688'8"	Dark gray shale
3688'8"	3691'7"	Laminated blue-gray and dark gray shale
3691'7"	3691'10"	Medium gray siltstone
3691'10"	3698'8"	Laminated blue-gray and dark gray shale
3698'8"	3698'11"	Dark gray laminated shale
3698'11"	3700'9"	Laminated blue-gray shale
3700'9"	3701'6"	Dark gray shale
3701'6"	3703'3"	Laminated blue-gray and dark gray shale
3703'3"	3704'6"	Blue-gray shale
3704'6"	3707'9"	Laminated blue-gray and dark gray shale
3707'9"	3711'1"	Blue-gray shale
3711'1"	3712'10"	Laminated blue-gray and dark gray shale
3712'10"	3712'11"	Light gray argillaceous siltstone
3712'11"	3716'	Blue-gray shale
3716'	3716'5"	Laminated dark gray and blue gray shale
3716'5"	3717'10"	Blue-gray shale
3717'10"	3718'2"	Laminated dark gray shale
3718'2"	3719'	Blue-gray shale
3719'	3720'	Laminated blue-gray and dark gray shale
3720'	3724'6"	Blue-gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3679'5"	3682'10"	Vertical, open
3684'	3690'11"	Vertical, open
3699'11"	3703'6"	Vertical, open
3718'8"	3718'9"	Folded zone with kerogen and pyrite fillings along a slickensided surface

Core #18 (3724'6"-3764'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3724'6"	3725'	Medium gray shale
3725'	3725'6"	Blue-gray shale
3725'6"	3725'9"	Blue-gray shale with one 2" black band
3725'9"	3726'	Blue-gray shale
3726'	3729'	Medium gray shale
3729'	3730'2"	Medium gray shale with four 1/4" light gray argillaceous siltstone layers
3730'2"	3730'10"	Blue-gray shale with black shale laminae and a 2" black band
3730'10"	3732'1"	Medium gray shale with a 2" light gray argillaceous siltstone
3732'1"	3732'11"	Blue-gray and black laminated shale
3732'11"	3733'1"	Blue-gray shale
3733'1"	3733'6"	Black shale with pyrite
3733'6"	3733'11"	Blue-gray shale
3733'11"	3735'5"	Medium gray shale
3735'5"	3736'2"	Blue-gray and black laminated shale
3736'2"	3736'8"	Blue-gray shale
3736'8"	3739'9"	Medium gray shale with one 1" silty band
3739'9"	3740'7"	Blue-gray shale with one 2" black band, black laminae
3740'7"	3743'	Medium gray shale
3743'	3743'4"	Blue-gray shale
3743'4"	3743'8"	Laminated blue-gray and black shale with two 2" black bands
3743'8"	3744'1"	Blue-gray shale
3744'1"	3746'	Medium gray shale with two 2" silty bands
3746'	3747'8"	Blue-gray and black laminated shale
3747'8"	3748'6"	Blue-gray shale
3748'6"	3748'9"	Blue-gray and black laminated shale
3748'9"	3750'8"	Blue-gray shale
3750'8"	3752'8"	Blue-gray and black laminated shale with four 2" black bands
3752'8"	3758'1"	Blue-gray shale, one 3" siltstone and one 3" black band with pyrite
3758'1"	3759'2"	Black shale with some blue gray laminae
3759'2"	3760'2"	Blue-gray shale
3760'2"	3761'11"	Medium gray shale
3761'11"	3762'3"	Black shale with pyrite
3762'3"	3763'10"	Medium gray shale
3763'10"	3764'2"	Black shale
3764'2"	3764'6"	Blue-gray shale

Core #18 (3724'6"-3764'6")

<u>From</u>	<u>To</u>	<u>Fractures</u>
3729'3"	3730'2"	Vertical, open
3730'4"	3731'1"	Vertical, open
3733'1"	3733'11"	Vertical, open
3734'3"	3736'3"	Vertical, open
3738'3"	3738'11"	Vertical, open
3762'9"	3763'	45° slickensided surface

Comment: Short core due to breakdown of booster.

- Core #19 (3764'6"-3799'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3764'6"	3766'4"	Medium gray shale
3766'4"	3766'8"	Black pyritic shale
3766'8"	3767'7"	Medium gray shale
3767'7"	3767'11"	Black pyritic shale
3767'11"	3768'6"	Medium gray calcareous shale
3768'6"	3771'	Medium gray slightly calcareous shale
3771'	3771'6"	Medium gray shale with calcareous silt-stone
3771'6"	3772'11"	Medium gray calcareous shale with one 2" black band
3772'11"	3773'9"	Medium gray and black banded shale (non-calcareous)
3773'9"	3774'1"	Contorted black pyritic shale
3774'1"	3776'	Black shale
3776'	3777'4"	Blue-gray shale
3777'4"	3779'	Blue-gray and black laminated shale with two blue-gray bands 2" and 3" thick
3779'	3781'5"	Blue-gray calcareous
3781'5"	3782'7"	Black and blue-gray laminated shale with one blue-gray band 1" thick
3782'7"	3784'2"	Blue-gray shale
3784'2"	3784'5"	Black with blue-gray laminated shale
3784'5"	3785'6"	Blue-gray shale
3785'6"	3787'2"	Medium gray shale
3787'2"	3787'5"	Medium gray calcareous shale
3787'5"	3796'5"	Medium gray shale with one 5" dark gray band and one 1/8" silt band
3796'5"	3797'9"	Medium gray shale, one 2" calcareous zone and two 1" calcareous zones
3797'9"	3799'6"	Blue-gray and black laminated shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3764'6"	3765'10"	Vertical, open
3767'2"	3767'10"	Vertical, open
3785'	3786'4"	Vertical, open
3786'4"	3787'4"	Vertical, open
3787'9"	3788'9"	Vertical, open
3793'2"	3794'5"	Vertical, open
3773'4"	3774'1"	Contorted zone with slickensides
3781'10"	3782'6"	Two 45° slickensided surfaces

Comment: Short core due to jamming of core barrel.



Core #20 (3799'6"-3858'4")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3799'6"	3806'9"	Blue-gray shale with black laminae
3806'9"	3808'2"	Black laminated shale
3808'2"	3809'11"	Blue-gray shale
3809'11"	3810'	Gray calcareous shale with randomly oriented doubly terminated calcite crystals
3810'	3810'3"	Blue-gray shale
3810'3"	3814'3"	Laminated black and blue-gray shale
3814'3"	3814'4"	Calcite crystal cluster
3814'4"	3815'6"	Blue-gray shale
3815'6"	3816'9"	Laminated black and blue-gray shale
3816'9"	3822'11"	Blue-gray shale
3822'11"	3823'9"	Black and blue-gray laminated shale
3823'9"	3824'3"	Blue-gray shale
3824'3"	3824'4"	Blue-gray calcareous shale
3824'4"	3830'6"	Medium gray shale
3830'6"	3831'4"	Laminated dark gray and medium gray shale
3831'4"	3835'5"	Medium gray shale with one 1/8" light gray silty layer
3835'5"	3835'6"	Medium gray calcareous shale
3835'6"	3837'	Medium gray shale with three 1/4" silty layers
3837'	3838'9"	Laminated black and medium gray shale
3838'9"	3840'9"	Medium blue-gray shale
3840'9"	3843'6"	Blue-gray shale with dark bands up to 3" thick
3843'6"	3845'5"	Blue-gray shale
3845'5"	3845'10"	Blue-gray calcareous shale
3845'10"	3850'1"	Medium gray shale with four 1/8" silty zones
3850'1"	3850'5"	Dark gray to black laminated shale
3850'6"	3851'6"	Blue-gray shale
3851'6"	3853'3"	Blue-gray and very dark gray laminated shale
3853'3"	3855'4"	Blue-gray shale
3855'4"	3856'9"	Grading through blue gray, medium gray, dark gray, to very dark gray laminated shale
3856'9"	3858'1"	Blue-gray shale
3858'1"	3858'4"	Dark gray shale

Fractures

No significant fractures

Core #21 (3858'4"-3916'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3858'6"	3859'1"	Black shale
3859'1"	3859'6"	Medium gray shale with black bands
3859'6"	3861'4"	Black shale
3861'4"	3862'11"	Medium gray green pyritic shale
3862'11"	3863'6"	Black shale
3863'6"	3866'4"	Medium gray shale
3866'4"	3866'6"	Light gray calcareous shale
3866'6"	3868'10"	Medium gray shale with seven 1/8" to 1/4" light gray silty streaks and pyrite blebs
3868'10"	3869'3"	Black shale with a large pyrite bleb
3869'3"	3871'5"	Medium gray shale
3871'5"	3871'11"	Black shale
3871'11"	3875'4"	Medium gray shale with two 1/2" silty layers, one slightly calcareous, black laminae
3875'4"	3875'9"	Black shale
3875'9"	3878'1"	Medium gray shale
3878'1"	3878'6"	Black shale
3878'6"	3882'5"	Medium gray shale with eight 1/4" silty zones
3882'5"	3882'11"	Black shale
3882'11"	3886'2"	Green, medium gray and dark gray laminated shale with six 1/4" silty layers, pyrite and thin calcareous laminae at top
3886'2"	3886'4"	Black shale with reworked top
3886'4"	3889'7"	Medium gray, green gray, and dark gray laminated shale with two 1/2" silty layers
3889'7"	3889'11"	Black shale
3889'11"	3893'	Medium gray, green gray and dark gray laminated shale
3893'	3893'5"	Black shale
3893'5"	3895'11"	Medium gray, dark gray, green gray laminated shale with two 1/2" silty layers at base (one slightly calcareous)
3895'11"	3897'2"	Medium gray, green gray, dark gray laminated shale
3897'2"	3897'5"	Black shale
3897'5"	3899'8"	Medium gray, green gray, dark gray laminated shale
3899'8"	3900'1"	Black shale
3900'1"	3902'7"	Black gray and dark gray laminated shale (mostly medium gray)
3902'11"	3904'7"	Medium gray, green gray, dark gray laminated shale
3904'7"	3904'11"	Black shale
3904'11"	3909'2"	Medium gray and dark gray shale -- darker near base, nine lighter calcareous layers
3909'2"	3910'8"	Black and dark gray laminated shale
3910'8"	3911'	Medium gray and dark gray laminated shale, numerous pyrite blebs
3911'	3912'1"	Medium gray shale with some dark laminae, one calcareous layer
3912'1"	3913'	Dark gray and black laminated shale
3913'	3915'3"	Medium gray and dark gray laminated shale with abundant pyrite at top

Core #21 (Cont'd)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3915'3"	3915'9"	Dark gray shale
3915'9"	3916'4"	Black shale
<u>From</u>	<u>To</u>	<u>Fractures</u>
3874'	3874'6"	Vertical, open mineralized

Core #22 (3916'6"-3956'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3916'6"	3916'9"	Black shale
3916'9"	3919'3"	Medium gray and dark gray laminated shale with pyrite in places
3919'3"	3919'8"	Black shale
3919'8"	3921'8"	Medium gray and dark gray laminated shale
3921'8"	3922'2"	Black shale
3922'2"	3924'6"	Medium gray and dark gray laminated shale
3924'6"	3925'10"	Black shale with several ½" dark bands
3925'10"	3926'10"	Dark gray laminated shale
3926'10"	3928'8"	Medium gray shale with dark gray laminae
3928'8"	3928'11"	Black shale
3928'11"	3934'7"	Medium gray and dark gray laminated shale, many silty layers to 1/8"
3934'7"	3934'10"	Black shale
3934'10"	3939'4"	Medium gray shale with dark gray laminae and silty layers to 1/8", one calcareous layer ½"
3939'4"	3940'1"	Medium gray shale
3940'1"	3942'2"	Medium gray and dark gray laminated shale, three silty streaks to ¼"
3942'2"	3942'4"	Black shale
3942'4"	3947'10"	Medium gray and dark gray shale, two 1/4" silty layers
3947'10"	3949'1"	Black shale
3949'1"	3952'6"	Medium and dark gray laminated shale with 1/16" pyrite bands near top
3952'6"	3953'2"	Black shale with pyrite blebs
3953'2"	3956'6"	Medium dark gray with darker laminae, pyrite bands to 1/4"

<u>From</u>	<u>To</u>	<u>Fractures</u>
3917'6"	3917'9"	45° slickensided surface
3924'6"	3925'4"	Mineralized slickensided surface
3935'6"	3936'6"	45° slickensided surface
3950'9"	3955'	Vertical, open
3952'3"	3953'2"	Mineralized slickensided surface

Comments: Short core due to jamming of core barrel.

## Core #23 (3956'6"-4007')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3956'6"	3958'4"	Medium and dark gray laminated shale
3958'4"	3959'4"	Black shale
3959'4"	3962'4"	Black and dark gray laminated shale with pyrite laminae
3962'4"	3965'9"	Medium and dark gray laminated shale with pyrite laminae
3965'9"	3966'	Black shale
3966'	3966'3"	Dark gray shale
3966'3"	3966'9"	Black shale
3966'9"	3970'9"	Dark gray shale and black shale with pyrite blebs and laminae
3970'9"	3972'10"	Black shale
3972'10"	3973'4"	Dark gray shale with pyrite at top and base
3973'4"	3976'8"	Black shale
3976'8"	3977'2"	Dark gray and black laminated shale
3977'2"	3977'7"	Black shale
3977'7"	3977'8"	Medium gray silty, pyritic calcareous band
3977'8"	3978'	Dark gray and black laminated shale
3978'	3980'1"	Black shale
3980'1"	3980'11"	Medium gray and green gray laminated shale with pyrite blebs and laminae
3980'11"	3981'	Dark gray very calcareous silty shale
3981'	3981'3"	Black shale
3981'3"	3981'10"	Medium gray and green gray shale with pyrite laminae and blebs
3981'10"	3982'6"	Black shale with pyrite laminae
3982'6"	3984'4"	Dark gray and black laminated shale with pyrite laminae
3984'4"	3985'6"	Black shale with some pyrite
3985'6"	3987'	Dark gray and black laminated shale
3987'	3987'7"	Dark gray shale with green bands
3987'7"	3993'2"	Black and dark gray shale with pyrite laminae
3993'2"	3997'7"	Black shale with pyrite blebs
3997'7"	3999'10"	Very dark gray shale with pyrite laminae and a 2" slightly calcareous layer at top
3999'10"	4000'2"	Black shale with two ½" pyrite bands
4000'2"	4000'5"	Dark gray shale
4000'5"	4002'8"	Black shale
4002'8"	4004'8"	Medium and black banded shale
4004'8"	4007'	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3977'3"	3979'11"	Vertical, open
3983'2"	3984'7"	Intensely crumbled zone
4000'	4003'4"	Intensely crumbled zone

Core #24 (4007'-4047'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
4007'	4014'3"	Black shale with pyrite laminae and vertical fillings, four calcareous bands and hairline calcite fillings
4014'3"	4014'6"	Calcareous, pyritic, silty layer with biotite (medium gray meta-bentonite
4014'6"	4016'3"	Folded and faulted black shale with pyrite laminae and dark calcareous shale band
4016'3"	4017'10"	Dark shale with many thin pyrite laminae
4017'10"	4019'7"	Black shale with several calcareous pyrite stringers, plus dark gray and green gray shale with calcareous pyrite layers to 1/8"
4019'7"	4025'1"	Very dark gray shale with calcareous pyrite bands
4025'1"	4026'6"	Laminated dark gray and gray green shale
4026'6"	4029'	Dark gray and black shale
4029'	4047'6"	Black shale (black in powder) with pyrite blebs and horizontal calcite veins along fractures

<u>From</u>	<u>To</u>	<u>Fractures</u>
4024'	4029'	Vertical, open fracture with slickensides
4028'7"	4029'9"	Mineralized vertical fracture
4036'	4036'10"	Vertical, open

## Core #25 (4047'6"-4056'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
4047'6"	4051'3"	Black shale
4051'3"	4051'4"	Calcareous, pyritic fossiliferous zone
4051'4"	4054'10"	Medium gray fossiliferous limestone
4054'10"	4054'11½"	Medium gray flint

<u>From</u>	<u>To</u>	<u>Fractures</u>
4047'6"	4051'3"	Vertical, open

Well No. 20402

Core #1 (2654' - 2712'8")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2654'	2682'3"	Gray green laminated shale with light gray siltstone layers showing cross-bedding up to 3" thick
2682'3"	2682'7"	Massive light gray siltstone
2682'7"	2703'5"	Green-gray laminated shale with light gray siltstone layers to 3" thick
2703'5"	2703'9"	Massive light gray siltstone
2703'9"	2712'8"	Green-gray laminated shale with light gray siltstone layers

Fractures

No open fractures



Core #2 (2712'8" - 2771'3")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2712'8"	2717'8"	Light gray laminated shale
2717'8"	2736'8"	Medium gray shale with dark gray laminae and argillaceous siltstone layers
2736'8"	2737'6"	Dark and medium gray shale layers about 2 1/2" thick
2737'6"	2742'8"	Medium gray shale with some dark gray laminae (less silty than above)
2742'8"	2770'8"	Medium and light gray laminated shale with light gray siltstone layers to 3" thick, and several dark shale layers up to 3" thick.
2770'8"	2771'3"	Dark gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2716'8"	2717'1"	Slightly mineralized fracture
2726'9"	2727'7"	Vertical, open
2733'	2733'7"	Vertical, open
2733'11"	2734'7"	Vertical, open
2751'4"	2751'7"	Vertical, open

## Core #3 (3000' - 3058'10")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3000'	3001'9"	Medium gray laminated shale with several 1/2" dark bands
3001'9"	3003'4"	Dark gray shale
3003'4"	3005'10"	Medium and dark gray banded shale
3005'10"	3006'4"	Medium gray shale with lighter calcareous shale laminae
3006'4"	3007'4"	Medium and dark gray banded shale
3007'4"	3007'11"	Medium and dark gray laminated shale, light gray calcareous shale, one 2" argillaceous siltstone
3007'11"	3009'1"	Medium gray shale
3009'1"	3010'	Dark gray shale
3010'	3013'10"	Medium and dark gray shale with a 1" contorted siltstone layer at base
3013'10"	3014'9"	Dark gray shale
3014'9"	3021'	Medium and dark gray banded shale with several 1/4" calcite layers
3021'	3022'1"	Dark gray shale
3022'1"	3026'10 1/2"	Medium dark gray shale with calcareous shale layers
3026'10 1/2"	3027'4"	Dark gray shale
3027'4"	3029'	Medium gray shale with a few dark gray bands
3029'	3035'6"	Dark gray shale with some medium gray bands and one 1/4" calcite layer
3035'6"	3036'10"	Medium gray shale
3036'10"	3038'1"	Dark gray shale
3038'1"	3039'1"	Medium and dark gray banded shale
3039'1"	3040'10"	Dark gray shale
3040'10"	3041'5"	Medium and dark gray banded shale
3041'5"	3042'5"	Dark gray shale
3402'5"	3044'6"	Medium and dark gray banded shale with calcareous zones to 1/2"
3044'6"	3047'5"	Dark gray shale
3047'5"	3050'2"	Medium and dark gray shale
3050'2"	3052'6"	Dark gray shale
3052'6"	3053'7"	Medium and dark gray shale with calcareous layers
3053'7"	3054'11"	Dark gray shale
3054'11"	3057'10"	Medium gray shale with dark bands, two 1" calcareous layers near base
3057'10"	3058'10"	Dark gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3006'8 1/2"	3012'4"	Vertical, mineralized, open
3021'	3022'2"	Vertical, open
3046'3"	3047'2"	Vertical, open
3049'10"	3050'7"	Vertical, open

## Core #4 (3058'10" - 3117'8")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3058'10"	3061'3"	Medium and dark gray banded shale, one calcareous layer near top
3061'3"	3062'	Medium gray shale
3062'	3063'1"	Dark gray shale
3063'1"	3064'2"	Medium and dark gray shale
3064'2"	3065'2 1/2"	Light gray calcareous shale
3065'2 1/2"	3065'7"	Dark gray shale
3065'7"	3065'9"	Light gray calcareous shale
3065'9"	3068'1"	Dark gray shale
3068'1"	3071'4"	Medium and dark gray banded shale, three calcareous zones
3071'4"	3073'	Light, medium and dark gray banded shale
3073'	3073'2"	Light gray calcareous shale
3073'2"	3075'6"	Light, medium and dark gray banded shale
3075'6"	3077'3"	Dark gray shale with some medium gray layers
3077'3"	3080'5"	Light to medium gray shale some 1/2" dark bands
3080'5"	3084'3"	Dark gray and medium gray banded shale
3084'3"	3087'6"	Medium gray shale, some dark bands to 1", basal calcareous zone
3087'6"	3088'1"	Light to medium gray shale with some dark bands, calcareous layers
3088'1"	3090'8"	Light to medium gray shale with several dark bands
3090'8"	3091'6"	Dark gray shale with medium gray bands
3091'6"	3098'4"	Light to medium and dark gray banded shale with two thin calcareous layers
3098'4"	3099'9"	Dark gray shale
3099'9"	3101'1"	Light to medium gray shale
3101'1"	3102'6"	Light, medium and dark gray banded shale
3102'6"	3103'9"	Light to medium gray shale with three calcareous layers
3103'9"	3104'8"	Light, medium and dark gray banded shale with several calcareous layers to 1" thick
3104'8"	3109'10"	Light and medium gray shale with several dark bands, two calcareous shale bands
3109'10"	3112'1"	Light and medium gray laminated shale
3112'1"	3115'2"	Dark gray shale with several light and medium gray bands
3115'2"	3117'8"	Dark and medium gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3059'8"	3064'10"	Vertical, open
3065'10"	3068'5"	Vertical, open also rubble zone
3069'10"	3071'2"	Vertical, open
3095'4"	3096'9"	Vertical, open
3098'2"	3099'3"	Vertical, open
3100'10"	3101'6"	Vertical, open
3112'2"	3112'9"	Vertical, open

## Core #5 (3298' - 3356'8")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3298'	3305'	Medium gray shale, several dark bands up to 3/4", eight calcareous layers
3305'	3305'4"	Light gray banded calcareous and dark gray shale
3305'4"	3306'	Medium and dark gray banded shale
3306'	3319'2"	Black shale with a 1" calcareous zone
3319'2"	3325'10"	Medium gray shale, several black bands
3325'10"	3325'11"	Light gray calcareous shale
3325'11"	3341'	Medium gray shale, black bands up to 3" some calcareous shale zones
3341'	3347'10"	Medium gray and black banded shale
3347'10"	3356'8"	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3301'	3301'8"	Vertical and open
3302'	3302'9"	Vertical and open
3304'2"	3305'	Vertical and open
3307'5"	3316'5"	Vertical and open
3316'7"	3317'10"	Vertical and open
3318'	3323'9"	Vertical and open
3324'5"	3325'6"	Vertical and open
3326'	3327'	Vertical and open
3337'4"	3339'	Vertical and open
3343'	3343'8"	Vertical and open
3345'5"	3346'7"	Vertical and open
3347'5"	3348'8"	Vertical and open
3349'	3356'8"	Vertical and open

Core #6 (3356'8" - 3414')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3356'	3357'	Black shale
3357'	3357'7"	Medium gray shale
3357'7"	3363'2"	Black shale
3363'2"	3365'3"	Black and medium gray banded shale
3365'3"	3367'6"	Medium gray shale, some black bands
3367'6"	3375'6"	Black shale, one large pyrite bleb
3375'6"	3376'	Medium gray shale
3376'	3379'11"	Black shale
3379'11"	3380'4"	Medium gray shale
3380'4"	3386'9"	Black shale
3386'9"	3388'10"	Medium gray shale
3388'10"	3389'5"	Black shale
3389'5"	3389'10"	Medium gray shale
3389'10"	3398'6"	Black shale, one 1/8" pyrite layer
3398'6"	3400'	Medium gray shale, one 2" siltstone bed
3400'	3400'4"	Black shale
3400'4"	3400'9"	Medium gray shale, one 1" light tan siltstone
3400'9"	3401'	Black shale
3401'	3406'9"	Light medium gray shale, two 1" calcareous bands, one 2" pyrite band
3406'9"	3409'5"	Black shale
3409'5"	3411'8"	Light-medium gray shale
3411'8"	3414'	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3356'	3357'	Vertical, open
3357'8"	3363'5"	Vertical, open
3364'4"	3380'	Vertical, open
3383'5"	3385'	Vertical, open
3390'	3398'11"	Vertical, open
3401'7"	3401'11"	Slickensided surface
3406'6"	3407'7"	Vertical, open
3409'	3410'5"	Vertical, open
3412'	3414'	Vertical, open

## Core #7 (3414' - 3471'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3414'	3416'	Black shale
3416'	3416'4"	Dark gray shale
3416'4"	3418'	Black shale, pyrite blebs
3418'	3419'2"	Dark gray shale
3419'2"	3419'8"	Black shale
3419'8"	3421'1"	Dark gray shale
3421'1"	3421'5"	Light gray calcareous siltstone
3421'5"	3422'9"	Medium and dark gray shale
3422'9"	3423'1 1/2"	Black shale
3423'1 1/2"	3424'1"	Medium gray shale, one 1 1/2" black band
3424'1"	3427'10"	Black shale
3427'10"	3428'5"	Black shale, 2 dark gray bands
3428'5"	3429'4"	Black shale
3429'4"	3431'4"	Medium gray shale, one 2" bed of very calcareous siltstone
3431'4"	3431'11"	Black shale
3431'11"	3434'6"	Medium gray shale
3434'6"	3434'7 1/2"	Light gray argillaceous limestone
3434'7 1/2"	3437'10"	Medium gray shale, one slightly calcareous 1" silt bed
3437'10"	3438'5"	Black shale
3438'5"	3439'4"	Medium gray shale
3439'4"	3439'8"	Light gray, slightly calcareous siltstone
3439'8"	3440'5"	Dark gray and black shale
3440'5"	3441'2"	Black shale
3441'2"	3442'	Medium gray and black shale (Turbidite layer)
3442'	3443'5"	Light gray, slightly calcareous, pyritic siltstone
3443'5"	3443'7"	Medium gray shale
3443'7"	3446'	Black shale, seven dark gray beds to 3"
3446'	3450'6"	Black shale, one large pyrite bleb
3450'6"	3450'11"	Very dark gray shale
3450'11"	3453'4"	Black shale
3453'4"	3455'9"	Banded dark gray and black shale
3455'9"	3458'6"	Medium gray shale
3458'6"	3462'3"	Black shale, numerous dark gray bands to 2" thick
3462'3"	3465'5"	Dark gray shale
3465'5"	3467'11"	Black shale
3467'11"	3468'1"	Medium gray shale
3468'1"	3471'4"	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3414'	3416'5"	Vertical, open
3416'6"	3419'11"	Vertical, open
3424'	3430'	Vertical, open

Core #7 (Cont'd)

<u>From</u>	<u>To</u>	<u>Fractures</u>
3434'6"	3434'7 1/2"	Slickensided surface
3437'10"	3438'	Vertical, open
3444'6"	3445'5"	Vertical, open
3453'2"	3455'8"	Vertical, open
3458'6"	3459'6"	Vertical, open
3461'6"	3465'6"	Vertical, open
3467'6"	3471'4"	Vertical, open

## Core #8 (3471'6" - 3531'6")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3474'	3477'2"	Black shale, one 2" medium gray shale band
3477'2"	3478'10"	Medium gray shale, three 1/2" black bands, one 1" black band
3478'10"	3479'7"	Black shale, one 1/2" medium gray shale band
3479'7"	3480'2"	Medium gray shale
3480'2"	3480'5"	Black shale with pyrite blebs
3480'5"	3481'10"	Medium gray shale with two pyritic black shale bands to 1"
3481'10"	3482'7"	Light to medium gray slightly calcareous shale
3482'7"	3486'7"	Medium gray calcareous shale with five 2-3" black bands
3486'7"	3486'9"	Light gray calcareous silty shale
3486'9"	3488'1"	Medium gray shale, three black shale bands 1" to 3" thick
3488'1"	3488'4"	Black shale
3488'4"	3488'9"	Medium gray shale
3488'9"	3490'5"	Medium gray and black banded shale
3490'5"	3499'8"	Black shale with several medium gray bands
3499'8"	3500'7"	Light to medium gray slightly calcareous silty shale, one black band
3500'7"	3500'11"	Black shale
3500'11"	3504'2"	Light to medium gray slightly calcareous silty shale with three pyritic black bands to 1 1/2"
3504'2"	3504'4"	Light gray calcareous siltstone
3504'4"	3505'5"	Black shale, two 1/4" medium gray bands
3505'5"	3505'8"	Medium gray shale
3505'8"	3505'11"	Light gray calcareous siltstone
3505'11"	3506'4"	Black and medium gray banded shale
3506'4"	3506'7"	Medium gray shale
3506'7"	3507'7"	Black and medium gray laminated shale (mostly black)
3507'7"	3515'	Black shale, four 1/4" medium gray bands, one 1/4" pyrite band
3515'	3515'3"	Light gray calcareous siltstone
3515'3"	3515'10"	Black shale with many medium gray bands
3515'10"	3516'3"	Medium gray slightly calcareous shale
3516'3"	3516'10"	Black and medium gray shale, (mostly black)
3516'10"	3517'2"	Medium gray and black banded shale, mostly medium gray
3517'2"	3518'9"	Black shale with some gray laminae
3518'9"	3519'	Medium gray shale, some black laminae
3519'	3525'6"	Black shale, some medium gray laminae and one 1/4" medium gray band
3526'7"	3527'7"	Black shale
3527'7"	3527'9"	Black and medium gray banded shale
3527'9"	3528'11"	Black shale, some pyrite blebs
3528'11"	3529'	Medium gray shale



Core #8 (Cont'd)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3529'	3529'5"	Black shale, some medium gray laminae
3529'5"	3530'	Light and medium gray banded slightly calcareous shale with one 1/4" black layer
3530'	3530'4"	Black shale
3530'4"	3531'	Light to medium gray banded shale (slightly calcareous), one 1/2" medium gray and black mottled shale band
3531'	3531'6"	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3470'	3475'3"	Vertical, open
3478'9"	3480'	Vertical, open
3480'	3481'3"	Vertical, open
3483'	3485'	Vertical, open
3487'	3488'	Vertical, open
3488'4"	3492'7"	Vertical, open
3492'7"	3494'1"	Vertical, open
3493'10"	3495'9"	Vertical, open
3495'8"	3496'8"	Vertical, open
3496'6"	3497'7"	Vertical, open
3501'	3503'2"	Vertical, open
3506'	3509'7"	Vertical, open
3509'11"	3512'	Vertical, open
3511'10"	3515'3"	Vertical, open
3517'7"	3518'9"	Vertical, open
3519'	3523'8"	Vertical, open

Core #9 (3531'6"-3591'3-1/2")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3531'6"	3532'6"	Black shale
3532'6"	3532'9"	Light gray shale with dark gray laminae
3532'9"	3533'3"	Black shale
3533'3"	3533'7"	Light to medium gray slightly calcareous shale
3533'7"	3538'3"	Black shale with several dark gray laminae and pyrite blebs
3538'3"	3538'5"	Medium to dark gray shale with contorted bedding
3538'5"	3545'10"	Black shale, two 1-1/2" contorted beds
3545'10"	3546'5"	Medium gray shale with dark gray laminae
3546'5"	3546'10"	Medium and black laminated shale
3546'10"	3547'7"	Medium and dark gray laminated shale
3547'7"	3553'2"	Black shale with dark gray bands and pyrite blebs
3553'2"	3554'1"	Black shale with pyrite blebs
3554'1"	3554'5"	Light gray to gray-green calcareous shale contorted beds
3554'5"	3554'10"	Dark gray and black laminated shale
3554'10"	3555'6"	Medium and dark gray banded shale
3555'6"	3557'9"	Black shale with pyrite blebs, two dark gray bands
3557'9"	3558'6"	Medium and dark gray laminated shale
3558'6"	3559'1"	Black shale
3559'1"	3559'8"	Medium and dark gray laminated shale
3559'8"	3560'6"	Black shale with some dark gray bands and abundant pyrite
3560'6"	3561'	Medium and dark gray laminated shale
3561'	3561'11"	Black shale with dark gray laminae and pyrite blebs
3561'11"	3562'7"	Light and dark gray laminated shale
3562'7"	3565'	Black shale
3565'	3565'5"	Slightly calcareous medium gray shale
3565'5"	3565'10"	Black shale
3565'10"	3566'5"	Medium gray slightly calcareous shale
3566'5"	3570'4"	Black shale with dark gray laminae
3570'4"	3571'	Medium and dark gray laminated shale
3571'	3572'5"	Black shale with pyrite blebs
3572'5"	3574'	Medium gray and black banded shale
3574'	3576'4"	Black shale
3576'4"	3576'9"	Medium gray shale
3576'9"	3576'10"	Light gray calcareous shale
3576'10"	3579'5"	Black shale with pyrite blebs and dark gray laminae
3579'5"	3580'7"	Medium and dark gray shale with pyrite laminae

Core #9 (Cont'd)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3580'7"	3582'11"	Black shale with dark gray zones and pyrite laminae
3582'11"	3583'4"	Medium and dark gray laminated shale
3583'4"	3584'8"	Black shale with three 1-1/2" medium gray bands
3584'8"	3585'	Medium gray shale
3585'	3586'	Black shale with two medium gray bands
3586'-	3586'5"	Medium gray shale with one black band and one 1/8" pyrite seam at 30° to the bedding
3586'5"	3590'3-1/2"	Black shale with one 3" medium gray band, several dark gray laminae and blebs

<u>From</u>	<u>To</u>	<u>Fractures</u>
3536'11"	3538'4"	Vertical, open
3549'4"	3550'10"	Vertical, open
3551'3"	3552'8"	Vertical, open
3559'7"	3561'4"	Vertical, open
3562'10"	3565'2"	Vertical, open
3566'8"	3571'	Vertical, open
3571'	3572'9"	Vertical, open
3573'6"	3574'11"	Vertical, open

Distinct Black Kerogen Layers

3533'11"  
3551'

## Core #10 (3892'-3950')

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3892'	3894'7"	Black shale with pyrite laminae and blebs
3894'7"	3895'2"	Medium and dark gray laminated shale with black shale and pyrite laminae
3895'2"	3897'	Black shale with three 3-4" bands of laminated shale
3897'	3899'5"	Black shale with pyrite blebs and laminae
3899'5"	3900'	Dark gray and black laminated shale with pyrite laminae
3900'	3903'10"	Black shale, some pyrite laminae and one bleb
3903'10"	3904'3"	Dark gray and black laminated shale with pyrite laminae
3904'3"	3907'	Black shale and pyrite laminae
3907'	3907'9"	Dark gray and black laminated shale, pyrite blebs
3907'9"	3911'6"	Black shale, many pyrite laminae, one 1" blebs
3911'6"	3912'	Medium gray shale with black laminae, one 2" pyrite bleb
3912'	3912'4"	Black shale
3912'4"	3915'	Medium and dark gray calcareous shale, one 3" black band
3915'	3920'	Black shale with pyrite laminae
3920'	3920'8"	Dark gray and black laminated shale
3920'8"	3925'9"	Black shale with some pyrite blebs and laminae
3925'9"	3928'6"	Very dark gray shale with black pyritic bands
3928'6"	3931'8"	Dark gray and black banded shale with pyrite bands and blebs
3931'8"	3934'1"	Black shale, some dark gray laminae and pyrite laminae
3934'1"	3934'7"	Medium and dark gray laminated shale
3934'7"	3937'	Black shale, two 4" medium-dark bands, pyrite
3937'	3937'5"	Dark gray and black laminated shale
3937'5"	3937'8"	Black shale
3937'8"	3940'3"	Black and dark gray laminated shale, pyrite laminae
3940'3"	3942'2"	Black and green-gray shale with one 3" light gray calcareous siltstone
3942'2"	3943'10"	Dark gray and black calcareous laminated shale with pyrite blebs and laminae
3943'10"	3944'6"	Intraformational breccia with clasts of varying sizes, calcareous fillings
3944'6"	3946'1"	Dark gray and black laminated shale, abundant pyrite
3946'1"	3947'6"	Medium gray-green shale with one 2" medium gray band

Core #10 (Cont'd)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3947'6"	3949'3"	Dark gray, black and green-gray laminated shale with some pyrite laminae
3949'3"	3949'10"	Medium gray very calcareous shale with a 1" contorted zone with calcite filling
3949'10"	3950'	Black shale

Fractures

3894'5"	Slickensided Surface
3894'9"	Slickensided surface
3898'3"	Slickensided surface
3924'	Slickensided surface
3924'11"	Slickensided surface
3927'3"	Slickensided surface
3929'8"	Slickensided surface
3931'2"	Slickensided surface
3934'6"	Slickensided surface
3942'2"	Calcite healed surface, 70° dip
3944'10"	Slickensided surface
3949'2"	Slickensided surface

Distinct Black Kerogen Layers

Observed at several depths in this core.

## Core #11 (3950'-3977'10")

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3950'	3954'1"	Very dark gray shale with eight calcareous pyritic bands to 1/8"
3954'1"	3958'2"	Very dark gray-brown very calcareous shale
3958'2"	3963'11"	Very calcareous black shale
3963'11"	3973'11"	Black shale
3973'11"	3974'1/2"	Calcareous pyritic silty zone with biotite (meta-bentonite)
3974'1/2"	3976'9"	Medium gray fossiliferous limestone

<u>From</u>	<u>To</u>	<u>Fractures</u>
3952'10"	3953'6"	45° slickensided surface with calcite
3955'2"	3956'5"	Several 45° slickensided surfaces
3972'9"	3972'10"	45° slickensided surface

Description of Cored Shale Intervals  
from Well Number 20336, Martin Co., KY

Run #1 (2432-2491)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2432	2491	Dark gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2434.3	2434.7	Vertical, parted
2435.3	2435.9	70° inclined, parted
2437.4	2438.4	Nearly vertical, parted
2438.6	2439.4	Nearly vertical, parted
2439.0	2439.4	Vertical, parted
2439.8	2440.8	Vertical, parted
2440.8	2442.6	Vertical, parted
2444.4	2445.3	Vertical, parted
2450.4	2451.2	Vertical, parted
2455.1	2460.6	Vertical, parted
2465.8	2465.5	Vertical, parted
2469.8	2470.5	Vertical, parted
2477.1	2477.9	70° inclined, parted
2481.0	2482	Nearly vertical, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
2434	2434	Thin calcareous seam
2454.8	2455.1	2 thin calcareous seams
2458	2458	Thin calcareous seam
2459.8	2459.8	Thin calcareous seam
2462.2	2462.7	4 thin calcareous seams
2468.0	2468.5	Numerous calcareous seams
2469.5	2470	Numerous calcareous seams
2475.2	2475.3	Numerous calcareous seams
2476.6	2477.0	3 calcareous seams
2478.0	2478.4	3 calcareous seams
2479.1	2479.2	Several calcareous seams
2487.2	2487.4	Calcareous zone
2488.5	2490	Scattered calcareous zones

Run #2 (2491-2550)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2491	2502	Dark gray shale
2502	2509	Medium dark gray shale
2509	2510	Medium gray and medium dark gray banded shale
2510	2511	Medium gray shale with dark gray bands
2511	2518	Medium gray shale
2518	2519	Medium gray and medium dark gray banded shale
2519	2526.9	Medium gray shale with dark gray bands
2526.9	2527.1	Medium gray siltstone
2527.1	2528.0	Medium dark gray shale
2528.0	2528.2	Medium gray siltstone
2528.2	2529.5	Medium gray shale with dark gray bands
2529.5	2529.9	Medium gray siltstone
2529.9	2530.2	Medium gray shale
2530.2	2530.3	Medium gray siltstone
2530.3	2532.2	Medium gray-green and dark gray banded shale
2532.2	2533.0	Medium gray-green shale
2533.0	2534.5	Medium gray-green shale with dark gray bands
2534.5	2537.0	Medium dark gray shale with medium gray green bands
2537.0	2539.2	Medium gray-green shale
2539.2	2540.5	Medium gray-green and dark gray banded shale
2540.5	2541.6	Dark gray shale
2541.6	2545.7	Medium gray-green shale with many silty bands up to .1' thick
2545.7	2545.9	Medium gray siltstone
2545.9	2550.0	Medium gray shale with three argillaceous siltstone bands

<u>From</u>	<u>To</u>	<u>Fractures</u>
2505.0	2505.8	Vertical fracture, parted
2511.4	2512.5	Vertical fracture, parted
2513.1	2513.9	Vertical fracture, parted
2514.5	2515.6	Vertical fracture, parted
2520.3	2521.2	Vertical fracture, parted
2524.8	2526.8	Vertical fracture, parted
2527.2	2528.0	Vertical fracture, parted
2528.6	2529.2	Vertical fracture, parted
2530.5	2531.5	Vertical fracture, parted
2533.5	2534.6	Vertical fracture, parted
2536.8	2540.4	Vertical fracture, parted
2546.3	2546.8	Vertical fracture, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
2491.0	2520.0	Numerous very thin light gray calcareous seams
2524.8	2524.8	Calcareous zone
2534.8	2535.0	Two thin calcareous seams
2536.3	2536.3	Thin calcareous seam
2541.0	2541.0	Thin calcareous seam
2545.3	2545.6	Vertical calcareous filling
2523.7	2523.7	Thin pyrite seam
2524.8	2525.0	Pyrite blebs



Run #3 (2550-2588.3)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2550.0	2553.9	Medium gray-green shale
2553.9	2554.5	Cross-bedded medium gray siltstone
2554.5	2556.3	Medium gray-green shale
2556.3	2556.9	Medium gray cross-bedded siltstone
2556.9	2567.9	Medium gray-green shale
2567.9	2577.6	Dark gray shale
2577.6	2581.7	Medium gray-green shale with some dark gray bands
2581.7	2582.4	Dark gray shale
2582.4	2586.5	Medium gray-green shale with dark gray bands
2586.5	2586.9	Dark gray shale
2586.9	2587.4	Medium gray-green shale
2587.4	2588.0	Dark gray shale
2588.0	2588.3	Medium gray-green shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2551.5	2552.4	Vertical, parted
2557.0	2558.2	Mineralized vertical fracture
2558.5	2559.0	Vertical fracture, parted
2560.8	2561.8	Vertical fracture, parted
2562.4	2563.4	Vertical fracture, parted
2563.6	2564.8	Vertical fracture, parted
2564.8	2565.5	Vertical fracture, parted
2567.6	2568.5	Vertical fracture, parted
2570.0	2570.6	Vertical fracture, parted
2571.2	2572.8	Vertical fracture, parted
2577.9	2578.6	Vertical fracture, parted
2579.8	2580.7	Vertical fracture, parted
2583.3	2583.9	Vertical fracture, parted
2584.2	2586.0	Vertical fracture, parted
2586.7	2587.7	Vertical fracture, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
2573.3	2573.5	Two calcareous bands
2576.7	2576.7	One calcareous band
2586.4	2586.4	One calcareous band

Run #4 (2588.3 to 2646.3)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2588.3	2590.6	Medium dark gray shale with dark gray bands
2590.6	2591.4	Dark gray shale
2591.4	2594.3	Medium dark gray shale
2594.3	2595.1	Dark gray shale
2595.1	2595.6	Medium gray shale
2595.6	2596.6	Medium dark gray shale
2596.6	2600.6	Medium dark gray shale with laminae and cross-beds of argillaceous silt
2600.6	2605.8	Medium gray-green shale
2605.8	2607.1	Dark gray shale
2607.1	2608.9	Medium gray-green shale with numerous dark gray bands
2608.9	2612.3	Dark gray shale
2612.3	2613.6	Medium gray-green shale
2613.6	2614.1	Dark gray shale
2614.1	2617.3	Medium gray-green shale with dark gray bands
2617.3	2618.3	Dark gray shale
2618.3	2619.9	Medium gray-green shale
2619.9	2620.5	Medium gray shale with light gray bands
2620.5	2626.1	Medium gray-green with medium gray bands
2626.1	2626.4	Medium gray, crossbedded, argillaceous siltstone
2626.4	2641.3	Medium gray-green shale with medium gray argillaceous siltstone bands up to .3' thick
2641.3	2642.7	Medium gray-green shale
2642.7	2646.3	Medium gray-green shale with numerous thin gray-green siltstone bands

<u>From</u>	<u>To</u>	<u>Fractures</u>
2589.3	2590.3	Vertical, parted
2591.2	2593.1	Vertical, parted
2593.3	2593.9	70° inclined, parted
2594.3	2596.6	Vertical, parted
2603.1	2605.1	Vertical, parted
2605.2	2606.0	Vertical, parted
2607.0	2608.3	Vertical, parted
2608.8	2609.4	Vertical, parted
2610.7	2611.6	Vertical, parted
2612.0	2612.8	Vertical, parted
2613.5	2619.9	Vertical, parted
2621.1	2621.9	Vertical, parted
2622.0	2623.5	Vertical, parted
2623.0	2626.2	Vertical, parted
2626.5	2627.2	Vertical, parted
2629.3	2630.4	Vertical, parted
2640.7	2643.5	Vertical, parted
2645.0	2646.3	Vertical, parted

Run #5 (2646.3 to 2704.8)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2646.3	2652.3	Medium gray-green shale with occasional medium dark gray laminae, and some argillaceous siltstone laminae
2652.3	2654.2	Black shale with argillaceous siltstone laminae
2654.2	2654.9	Medium gray-green shale
2654.9	2658.0	Black shale with several dark gray laminae
2658.0	2659.1	Black shale
2659.1	2659.4	Medium gray shale with one thin dark gray band
2659.4	2670.7	Black shale with numerous dark gray laminae
2670.7	2671.6	Medium gray-green shale with black laminae
2671.6	2680.1	Black shale with occasional medium to dark gray laminae
2680.1	2680.4	Medium gray-green shale with black laminae
2680.4	2683.1	Black shale
2683.1	2683.3	Medium dark gray shale with very fine pyrite laminae
2683.3	2692.0	Black shale with scattered medium gray laminae
2692.0	2692.1	Medium gray green shale
2692.1	2698.5	Black shale with occasional medium gray laminae
2698.5	2698.6	Medium gray green and black laminated shale
2698.6	2704.8	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2647.0	2649.0	Vertical, parted
2649.2	2650.7	Vertical, parted
2650.9	2651.7	Vertical, parted
2653.0	2656.0	Vertical, parted
2656.0	2657.5	Vertical, parted
2657.9	2660.7	Vertical, parted
2660.9	2661.6	Vertical, parted
2661.8	2662.7	Vertical, parted
2662.7	2664.4	Vertical, parted
2667.8	2668.5	Vertical, parted
2668.5	2671.5	Vertical, parted
2671.6	2674.3	Vertical, parted
2674.4	2677.7	Vertical, parted
2677.7	2677.7	Slickensided horizontal plane
2678.6	2680.9	Slickensided horizontal plane
2680.9	2681.9	Slickensided horizontal plane
2682.0	2683.9	Slickensided horizontal plane
2684.0	2686.6	Slickensided horizontal plane
2687.0	2687.9	Slickensided horizontal plane
2688.0	2688.9	Slickensided horizontal plane
2688.9	2691.7	Slickensided horizontal plane
2692.5	2693.3	Slickensided horizontal plane
2695.5	2697.6	3 nearly vertical, parted, overlapping fractures
2698.3	2698.5	Inclined slickensided surface
2701.0	2704.3	5 nearly-vertical, curved parted surfaces

Run #5 (Cont.) (2646.3 to 2704.8)

<u>From</u>	<u>To</u>	<u>Other Features</u>
2655.0	2670.0	Scattered pyrite laminae and blebs
2670.0	2685.0	Occasional pyrite laminae
2697.3	2697.3	Large pyrite bleb

Run #6 (2704.8-2762.8)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2704.8	2721.4	Black shale
2721.4	2728.7	Medium dark gray shale with black laminae
2728.7	2735.8	Black shale
2735.8	2741.7	Medium dark gray shale with black laminae
2741.7	2744.7	Black shale with dark gray laminae
2744.7	2749.2	Black shale with several dark gray laminae
2749.2	2762.8	Black shale with numerous dark gray laminae

<u>From</u>	<u>To</u>	<u>Fractures</u>
2704.8	2705.7	Vertical, parted
2708.5	2711.9	Vertical, parted
2711.9	2725.6	Vertical, parted
2725.6	2725.6	Horizontal slickensided plane
2725.6	2726.4	Vertical, parted
2727.4	2728.2	Vertical, parted
2728.1	2730.6	Vertical, parted
2730.7	2732.5	Vertical, parted
2733.2	2734.0	Vertical, parted
2734.1	2734.9	Vertical, parted
2736.2	2737.6	Vertical, parted
2738.0	2738.8	Vertical, parted
2740.1	2741.4	Vertical, parted
2741.5	2748.8	Vertical, parted
2762.0	2762.8	Vertical, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
2718.8	2719.0	Pyrite blebs

Run #7 (2762.8-2820.6)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2762.8	2773.5	Black shale with medium dark gray laminae
2773.5	2773.6	Medium gray shale
2773.6	2776.8	Black shale
2776.8	2777.0	Medium gray shale
2777.0	2780.2	Black shale
2780.2	2783.3	Medium gray shale with dark gray laminae
2783.3	2786.5	Black shale with dark gray laminae
2786.5	2786.8	Medium dark gray shale
2786.8	2792.4	Black shale
2792.4	2800.7	Medium dark gray shale with black bands
2800.7	2801.0	Medium gray argillaceous siltstone
2801.0	2802.9	Medium dark gray shale
2802.9	2803.2	Medium gray silty shale
2803.2	2803.5	Medium dark gray shale
2803.5	2804.3	Medium gray shale
2804.3	2804.4	Medium gray silty shale
2804.4	2805.1	Medium dark gray shale
2805.1	2805.3	Medium gray silty shale
2805.3	2809.7	Medium dark gray shale with thin black layers
2809.7	2810.2	Medium gray shale
2810.2	2810.4	"Loading feature" dividing medium gray and medium dark gray shale
2810.4	2811.7	Medium dark gray shale with thin black layers
2811.7	2812.2	Medium gray silty shale
2812.2	2817.3	Medium dark gray shale with thin black and medium gray brown layers
2817.3	2818.0	Medium gray silty shale
2818.0	2819.3	Medium dark gray shale
2819.3	2819.6	Black shale
2819.6	2820.0	Medium gray slightly silty shale
2820.0	2820.6	Medium dark gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2763.0	2768.2	Vertical, parted
2769.4	2770.7	Vertical, parted
2774.0	2775.7	Vertical, parted
2777.4	2778.5	Vertical, parted
2779.6	2782.1	Vertical, parted
2787.0	2788.9	Vertical, parted
2789.5	2790.3	Vertical, parted
2790.4	2795.1	Vertical, parted
2796.7	2797.3	Vertical, parted
2797.6	2799.3	Vertical, parted
2801.3	2802.9	Vertical, parted
2812.2	2813.5	Vertical, parted
2814.4	2816.4	Vertical, parted
2816.2	2817.0	Vertical, parted
2817.8	2819.8	Vertical, parted

Run #7 (Cont.) 2762.8-2820.6)

<u>From</u>	<u>To</u>	<u>Other Features</u>
2764.0	2772.0	Pyrite laminae and one large bleb
2797.0		Two pyrite laminae

Run #8 (2820.6-2879.3)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2820.6	2823.6	Medium dark gray shale with thin black bands
2823.6	2824.3	Black shale with medium dark gray laminae
2824.3	2829.0	Medium dark gray shale
2829.0	2829.2	Medium gray and medium dark gray laminated shale
2829.2	2830.3	Medium dark gray shale
2830.3	2830.5	Medium gray shale
2830.5	2830.9	Black shale with medium dark gray laminae
2830.9	2831.9	Medium dark gray shale with three medium dark to black bands
2831.9	2833.5	Medium dark gray shale
2833.5	2833.7	Medium gray silty shale
2833.7	2834.1	Medium dark gray shale
2834.1	2834.4	Medium gray shale with medium gray green laminae
2834.4	2838.7	Medium dark gray shale with several black bands
2838.7	2839.0	Medium gray shale
2839.0	2840.8	Medium dark gray shale
2840.8	2840.9	Medium gray silty shale
2840.9	2841.4	Medium dark gray shale
2841.4	2841.5	Medium gray silty shale
2841.5	2841.9	Medium dark gray shale
2841.9	2842.2	Medium gray silty shale
2842.2	2843.1	Medium dark gray shale
2843.1	2843.5	Black shale with medium dark gray laminae
2843.5	2844.9	Medium dark gray shale with black bands
2844.9	2845.3	Black shale
2845.3	2845.7	Medium dark gray shale
2845.7	2846.2	Black shale with medium gray laminae
2846.2	2846.5	Medium dark gray
2846.5	2846.7	Black shale
2846.7	2847.1	Medium dark gray shale
2847.1	2847.4	Black shale with medium dark gray laminae
2847.4	2848.1	Medium dark gray shale
2848.1	2848.9	Black shale with medium dark gray laminae
2848.9	2849.7	Medium dark gray shale
2849.7	2850.0	Medium dark gray and medium gray banded shale
2850.0	2850.7	Medium dark gray shale
2850.7	2851.0	Medium gray silty shale
2851.0	2851.3	Medium dark gray shale
2851.3	2851.5	Medium dark gray and dark gray laminated shale
2851.5	2851.8	Medium dark gray shale
2851.8	2852.0	Medium gray shale
2852.0	2852.2	Medium dark gray shale
2852.2	2852.3	Dark gray shale
2852.3	2852.7	Medium dark gray shale
2852.7	2855.8	Medium dark gray shale with dark gray and black bands
2855.8	2856.9	Dark gray and medium dark gray laminated shale
2856.9	2857.9	Medium dark gray shale with dark gray bands
2857.9	2858.5	Dark gray and medium dark gray laminated shale
2858.5	2861.5	Medium dark gray shale with dark gray bands
2861.5	2862.6	Dark gray shale with medium dark gray laminae
2862.6	2862.9	Medium dark gray shale



Run #8 (Cont.) (2820.6-2879.3)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2862.9	2864.1	Dark gray and medium dark gray laminated shale
2864.1	2865.1	Black shale
2865.1	2866.5	Medium dark gray shale
2866.5	2867.5	Black shale with medium dark gray laminae
2867.5	2867.9	Medium dark gray shale
2867.9	2879.3	Black shale with medium dark gray laminae

<u>From</u>	<u>To</u>	<u>Fractures</u>
2820.6	2822.0	Vertical, parted
2821.8	2824.6	Vertical, parted
2824.7	2825.7	Vertical, parted
2825.7	2829.0	Vertical, parted
2829.0	2830.4	Vertical, parted
2831.9	2833.5	Vertical, parted
2834.2	2838.6	Vertical, parted
2839.2	2842.2	Vertical, parted
2842.2	2847.5	Vertical, parted
2847.9	2855.3	Vertical, parted
2855.4	2860.0	Vertical, parted
2859.9	2862.7	Vertical, parted
2863.6	2864.9	Vertical, parted
2865.1	2866.3	Vertical, parted
2867.8	2869.2	Parted, curved, surface
2869.2	2870.6	Parted, curved, surface
2871.0	2875.3	Vertical, parted
2876.0	2879.3	Vertical, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
2869.8	2869.8	Pyrite bleb
2866.3	2866.3	Carbonate layer

Run #9 (2879.3 to 2939.0)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2879.3	2893.6	Black shale with medium and dark gray laminae
2893.6	2893.8	Medium gray shale
2893.8	2894.6	Black shale with medium and dark gray bands
2894.6	2896.5	Medium gray shale with dark gray and black bands
2896.5	2898.3	Black shale with dark gray bands
2898.3	2899.1	Medium dark gray shale
2899.1	2899.5	Black shale with medium dark gray laminae
2899.5	2901.7	Medium dark gray shale with occasional black bands
2901.7	2902.1	Black shale
2902.1	2902.4	Medium dark gray shale
2902.4	2902.7	Black shale
2902.7	2903.3	Black shale with medium dark gray laminae
2903.3	2903.8	Medium dark gray shale
2903.8	2904.0	Black shale
2904.0	2904.3	Medium dark gray shale
2904.3	2904.5	Black shale
2904.5	2904.8	Medium dark gray shale
2904.8	2905.0	Black shale
2905.0	2906.0	Medium dark gray shale
2906.0	2906.1	Dark gray shale
2906.1	2906.5	Medium dark gray shale
2906.5	2906.9	Dark gray shale
2906.9	2907.1	Medium dark gray shale
2907.1	2907.3	Dark gray shale
2907.3	2908.3	Medium dark gray shale with occasional dark gray bands
2908.3	2920.5	Black shale with medium dark gray laminae
2920.5	2920.7	Medium gray and black laminated shale divided by "loading structure"
2920.7	2928.2	Black with medium gray laminae
2928.2	2934.7	Medium dark gray shale with black bands and occasional slightly silty medium gray bands
2934.7	2934.8	Medium gray argillaceous siltstone
2934.8	2936.7	Medium dark gray shale with occasional black bands
2936.7	2937.1	Black shale
2937.1	2939.0	Medium dark gray shale with occasional black bands

<u>From</u>	<u>To</u>	<u>Fractures</u>
2879.3	2880.1	Vertical, parted
2882.0	2883.4	Vertical, parted
2884.0	2886.8	Vertical, parted
2886.8	2886.8	Mineralized horizontal fracture
2886.8	2888.2	Vertical, parted
2888.4	2889.2	Vertical, parted
2889.6	2899.1	Vertical, parted
2901.4	2902.2	Vertical, parted

Run #9 (Cont.) (2879.3 to 2939.0)

<u>From</u>	<u>To</u>	<u>Fractures</u>
2902.4	2904.3	Vertical, parted
2904.3	2904.3	Mineralized horizontal fracture
2904.3	2905.7	Vertical, parted
2905.5	2906.3	Vertical, parted
2906.4	2909.3	Vertical, parted
2911.2	2913.5	Vertical, parted
2914.1	2918.9	Vertical, parted
2929.0	2930.2	Vertical, parted
2930.3	2932.4	Vertical, parted
2933.0	2934.0	Vertical, parted
2935.7	2939.0	Vertical, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
2894.0	2894.4	Pyrite blebs and layers
2900.8	2900.8	Pyrite layer
2913.5	2913.9	Pyrite blebs
2916.5	2916.5	Pyrite blebs
2919.7	2919.7	Pyrite blebs
2934.0	2934.0	Pyrite blebs

Run #10 (2939.0-2997.5)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2939.0	2940.0	Medium dark gray shale with black bands
2940.0	2940.8	Black shale
2940.8	2941.3	Medium dark gray shale
2941.3	2941.9	Black shale
2941.9	2942.2	Medium dark gray shale
2942.2	2943.1	Black shale
2943.1	2943.5	Medium dark gray shale
2943.5	2948.5	Black shale with several medium gray and medium dark gray bands
2948.5	2951.1	Medium dark gray with several black bands
2951.1	2951.8	Black shale
2951.8	2953.5	Medium gray shale with black bands
2953.5	2953.6	Medium gray silty shale
2953.6	2954.5	Medium dark gray shale with several black bands
2954.5	2954.6	Medium gray silty shale
2954.6	2955.9	Medium dark gray with several dark bands
2955.9	2956.0	Medium gray silty shale
2956.0	2958.0	Medium dark gray shale with several black bands
2958.0	2962.3	Black shale
2962.3	2962.4	Medium dark gray shale
2962.4	2976.7	Black shale
2976.7	2976.8	Medium dark gray shale
2976.8	2977.2	Black shale
2977.2	2977.5	Medium dark gray shale with several distorted black laminae
2977.5	2978.3	Black shale
2978.3	2980.0	Alternating bands of black and medium dark gray shale (many at 70° to 80° from vertical)
2980.0	2980.6	Black shale
2980.6	2981.4	Medium dark gray shale with several distorted black laminae
2981.4	2982.0	Black shale
2982.0	2982.6	Medium dark gray shale
2982.6	2982.7	Black shale
2982.7	2983.3	Medium dark gray shale
2983.3	2983.7	Black shale
2983.7	2984.7	Medium dark gray shale
2984.7	2985.4	Black shale
2985.4	2985.9	Medium gray shale
2985.9	2987.1	Black shale
2987.1	2987.2	Medium dark gray shale
2987.2	2990.4	Black shale
2990.4	2990.6	Medium dark gray shale
2990.6	2997.5	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2939.0	2939.6	Vertical, parted
2941.0	2943.6	Vertical, parted
2946.6	2948.6	Vertical, parted

## Run #10 (2939.0-2997.5) (Cont.)

<u>From</u>	<u>To</u>	<u>Fractures</u>
2948.6	2952.0	Vertical, parted
2954.6	2954.6	Mineralized horizontal joint
2957.0	2957.9	Vertical, parted
2958.4	2959.6	Vertical, parted
2963.0	2964.2	Vertical, parted
2966.8	2968.6	Vertical, parted
2968.6	2978.4	Vertical, parted
2978.3	2980.8	Vertical, parted
2980.8	2990.0	2 Slickensided 45° joints
2981.0	2983.4	2 Slickensided 45° joints
2988.2	2989.0	2 Slickensided 45° joints
2989.0	2990.4	2 Slickensided 45° joints
2991.5	2993.8	70° inclined, parted
2993.6	2997.5	70° inclined, parted
<u>From</u>	<u>To</u>	<u>Fractures</u>
2939.0	2942.0	Abundant pyrite blebs
2944.0	2948.0	Scattered pyrite blebs
2958.0	2978.0	Pyrite blebs
2978.5	2978.6	Large pyrite nodule
2981.5	2987.0	Pyrite blebs in black shale zones
2991.5	2997.0	Large pyrite nodule and blebs
2988.7	2988.7	Discrete black organic substance

Run #11 (2997.5-3047.8)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
2997.5	3001.4	Medium dark gray shale with medium gray and black bands
3001.4	3002.2	Black shale
3002.2	3002.3	Medium dark gray and black shale separated by "loading structure"
3002.3	3002.8	Medium dark gray shale
3002.8	3004.2	Medium gray shale
3004.2	3004.7	Black shale with medium dark gray laminae
3004.7	3007.8	Medium gray shale
3007.8	3008.3	Black shale
3008.3	3009.0	Medium dark gray and medium gray shale with "loading feature"
3009.0	3009.3	Medium dark gray shale
3009.3	3011.6	Very dark gray shale
3011.6	3011.9	Dark gray and medium dark gray shale with "loading structure"
3011.9	3014.0	Black shale
3014.0	3014.1	Medium gray shale
3014.1	3016.3	Black shale
3016.3	3016.6	Medium dark gray shale with dark gray and black bands showing loading structure
3016.6	3017.1	Black shale
3017.1	3017.4	Black shale with medium dark gray bands and "loading structure"
3017.4	3018.2	Black shale
3018.2	3018.7	Black shale with medium dark gray band showing "loading structure"
3018.7	3019.4	Black shale
3019.4	3019.5	Medium gray shale
3019.5	3019.8	Black shale
3019.8	3020.5	Medium dark gray shale with medium gray and black layers showing "loading structure"
3020.5	3022.3	Medium dark gray and black banded shale
3022.3	3022.5	Black and medium dark gray shale with "loading structure"
3022.5	3022.8	Black shale
3022.8	3022.9	Medium dark gray and black laminated shale
3022.9	3024.0	Black shale
3024.0	3024.1	Dark gray and black shale with contorted bedding
3024.1	3025.2	Black shale
3025.2	3026.2	Medium gray shale
3026.2	3026.5	Medium gray and dark gray shale with contorted bedding
3026.5	3027.1	Black shale
3027.1	3029.6	Medium gray shale
3029.6	3029.8	Medium dark gray and black banded shale
3029.8	3030.5	Black shale
3030.5	3031.0	Medium dark gray shale with several black bands
3031.0	3031.5	Black shale
3031.5	3031.6	Medium dark gray shale
3031.6	3031.9	Black shale
3031.9	3032.3	Medium gray shale
3032.3	3032.4	Black shale

Run #11 (2997.8-3047.8) (Cont.)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3032.4	3033.1	Medium gray shale
3033.1	3033.3	Black, medium gray, and medium dark gray banded shale with contorted bedding black bands at 60° to vertical
3033.3	3034.2	Medium gray and medium dark gray banded shale with fine black, irregular streaks, contorted bedding
3034.2	3036.8	Medium gray shale
3036.8	3037.6	Medium gray calcareous shale
3037.6	3038.5	Black shale
3038.5	3038.9	Medium dark gray shale
3038.9	3039.2	Black shale
3039.2	3039.6	Medium dark gray shale
3039.6	3040.8	Black shale
3040.8	3040.9	Medium dark gray shale
3040.9	3043.2	Black shale
3043.2	3043.3	Medium dark gray and black shale with contorted bedding
3043.3	3047.8	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
2997.5	3000.0	Vertical, parted
3000.0	3000.8	Vertical, parted
3001.6	3007.5	Vertical, parted
3007.7	3008.9	Vertical, parted
3009.1	3014.7	Vertical, parted
3015.3	3019.8	Vertical, parted
3020.7	3026.1	Vertical, parted
3026.1	3027.2	Vertical, parted
3027.2	3036.3	Vertical, parted
3037.6	3047.8	

<u>From</u>	<u>To</u>	<u>Other Features</u>
3112.0	3114.0	Pyrite blebs
3120.4	3120.5	Large pyrite nodule
3122.8	3124.0	Pyrite blebs
3033.1	3033.3	Pyrite blebs
3041.3	3041.4	Large pyrite nodule
3036.8	3037.5	Calcareous zone

Run #12 (3047.8-3105.3)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3047.8	3049.4	Black shale with medium dark gray laminae
3049.4	3049.5	Medium dark gray shale
3049.5	3057.9	Medium dark gray shale with black and medium gray laminae
3057.9	3058.0	Medium dark gray and dark gray shale with contorted bedding
3058.0	3063.1	Medium dark gray shale with black laminae
3063.1	3064.9	Black shale
3064.9	3066.0	Medium dark gray and black shale with contorted bedding
3066.0	3067.7	Black shale
3067.7	3068.1	Medium dark gray and dark gray shale with contorted bedding
3068.1	3069.2	Black shale with occasional gray laminae
3069.2	3069.3	Black and medium dark gray shale with contorted bedding
3069.3	3075.7	Black shale
3075.7	3076.1	Medium dark gray and dark gray laminated shale
3076.1	3078.7	Black shale
3078.7	3080.4	Medium gray shale with some dark gray bands
3080.4	3080.6	Medium dark gray shale
3080.6	3083.1	Medium dark gray and dark gray laminated shale
3083.1	3084.2	Black shale with dark gray laminae
3084.2	3084.9	Medium dark gray and dark gray laminated shale
3084.9	3085.1	Black shale
3085.1	3085.7	Medium dark gray and dark gray laminated shale
3085.7	3088.3	Black shale
3088.3	3088.5	Medium dark gray and black shale with contorted bedding
3088.5	3088.9	Black shale
3088.9	3089.2	Medium dark gray and dark gray laminated shale
3089.2	3089.7	Black shale
3089.7	3089.9	Medium dark gray shale
3089.9	3090.7	Black shale
3090.7	3090.8	Medium dark gray shale
3090.8	3092.1	Black shale
3092.1	3092.3	Medium dark gray and black shale with contorted bedding
3092.3	3092.6	Black shale
3092.6	3092.8	Medium dark gray and dark gray laminated shale
3092.8	3101.7	Black shale
3101.7	3102.1	Medium dark gray and dark gray laminated shale
3102.1	3102.7	Black shale with dark gray laminae
3102.7	3102.9	Medium dark and dark gray laminated shale
3102.9	3105.3	Black shale



<u>From</u>	<u>To</u>	<u>Fractures</u>
3047.8	3049.1	Vertical, parted
3049.7	3053.5	Vertical, parted
3053.9	3058.2	Vertical, parted
3058.3	3059.6	Vertical, parted
3063.6	3065.0	Vertical, parted
3065.3	3066.0	Vertical, parted
3066.0	3067.0	Vertical, parted
3068.8	3069.6	Vertical, parted
3070.0	3071.6	Vertical, parted
3071.8	3082.8	Vertical, parted
3083.1	3084.2	Vertical, parted
3085.7	3089.8	Vertical, parted
3090.0	3092.2	Vertical, parted
3092.8	3098.8	Vertical, parted
3100.7	3104.6	Vertical, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
3047.8	3049.0	Pyrite blebs
3066.0	3067.3	Pyrite blebs
3068.4	3068.5	Two pyrite laminae
3070.5	3074.0	Pyrite blebs
3086.0	3087.8	Pyrite blebs
3094.1	3094.1	Pyrite laminae
3195.9	3095.9	Pyrite blebs
3097.4	3097.4	Pyrite blebs
3090.0		Calcareous band

Run #13 (3105.3-3164.1)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3105.3	3105.8	Dark gray and medium dark gray laminated shale with contorted bedding
3105.8	3118.6	Black shale with several dark gray laminae
3118.6	3119.0	Medium gray green and dark gray laminated shale
3119.0	3119.2	Black and dark gray laminated shale
3119.2	3120.1	Medium gray-green and dark gray laminated shale
3120.1	3120.8	Black shale
3120.8	3122.0	Medium gray-green shale
3122.0	3123.2	Black shale
3123.2	3130.2	Medium gray-green shale with occasional dark gray laminae
3130.2	3130.6	Medium gray argillaceous siltstone
3130.6	3135.9	Medium gray-green shale with numerous dark group laminae
3135.9	3136.2	Black shale with some gray-green laminae
3136.2	3149.0	Medium dark gray-green shale with dark gray laminae
3149.0	3151.5	Medium gray argillaceous siltstone
3151.5	3156.4	Medium gray-green shale with dark gray laminae
3156.4	3157.2	Black shale
3157.2	3158.7	Medium dark gray-green shale with dark gray laminae
3158.7	3159.9	Medium gray-green slightly silty shale
3159.9	3162.7	Medium dark gray-green shale with dark gray laminae
3162.7	3164.1	Medium dark gray-green slightly silty shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3105.8	3108.5	Vertical, parted
3108.4	3111.6	Vertical, parted
3112.0	3116.5	Vertical, parted
3116.5	3117.6	Vertical, parted
3117.6	3121.3	Vertical, parted
3119.8	3120.3	Slickensided surface 40° from vertical
3121.7	3123.7	Vertical, parted
3126.5	3129.2	Vertical, parted
3129.2	3140.2	Vertical, parted
3140.4	3141.6	Vertical, parted
3141.7	3142.5	Vertical, parted
3142.7	3143.5	Vertical, parted
3143.9	3144.5	Vertical, parted
3144.4	3147.3	Vertical, parted
3147.5	3150.4	Vertical, parted

Run #13 (3105.3-3164.1) cont'd.

<u>From</u>	<u>To</u>	<u>Fractures</u>
3150.8	3151.3	Mineralized vertical fracture
3151.7	3158.4	Vertical, parted
3159.6	3161.2	Vertical, parted
3161.5	3162.2	Vertical, parted
<u>From</u>	<u>To</u>	<u>Other Features</u>
3122.8	3122.9	Calcareous horizontal filling
3123.2	3130.2	Slightly calcareous zone
3149.0	3151.5	Calcareous zone
3106.0	3108.0	Pyrite blebs
3117.5	3117.5	Pyrite nodule
3158.4	3158.5	Pyrite blebs
3154.0	3154.0	Discrete black organic layer
3156.6	3156.6	Discrete black organic layer

Run #14 (3164.1-3221.0)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3164.1	3168.5	Medium gray green with dark gray laminae
3168.5	3169.0	Dark gray shale
3169.0	3172.1	Medium gray green shale
3172.1	3172.4	Dark gray shale
3172.4	3174.7	Dark gray-green shale
3174.7	3175.3	Medium gray-green shale
3175.3	3175.6	Medium gray shale
3175.6	3178.2	Medium gray-green shale
3178.2	3178.4	Very dark gray shale
3178.4	3183.0	Medium gray-green shale
3183.0	3186.2	Light gray-green shale
3186.2	3187.1	Very dark gray shale
3187.1	3188.4	Medium gray-green shale
3188.4	3188.7	Light gray-green shale
3188.7	3189.7	Medium gray-green shale
3189.7	3189.9	Very dark gray shale
3189.9	3192.4	Medium gray-green shale
3192.4	3192.5	Very dark gray shale
3192.5	3193.1	Medium gray-green shale
3193.1	3193.9	Light gray-green shale
3193.9	3197.1	Medium gray-green shale
3197.1	3198.5	Medium gray-green shale with dark gray laminae
3198.5	3198.9	Medium gray-green shale
3198.9	3199.1	Very dark gray shale
3199.1	3201.1	Medium gray-green shale
3201.1	3202.1	Medium gray-green and dark gray laminated shale
3202.1	3205.5	Medium gray-green shale
3205.5	3206.3	Very dark gray shale
3206.3	3208.8	Medium gray-green shale
3208.8	3208.9	Very dark gray shale
3208.9	3209.3	Medium gray-green shale
3209.3	3210.3	Medium gray shale
3210.3	3210.5	Dark gray shale
3210.5	3211.0	Medium gray shale
3211.0	3211.4	Light gray argillaceous siltstone
3211.4	3212.0	Medium dark gray shale
3212.0	3212.4	Medium dark gray shale
3212.4	3214.4	Medium gray shale
3214.4	3214.5	Dark gray shale
3214.5	3216.7	Medium gray shale
3216.7	3218.2	Black shale
3218.2	3221.0	Medium gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3181.7	3182.8	Vertical, parted
3183.8	3185.5	Vertical, parted
3186.0	3187.9	Vertical, parted

Run #14 (3164.1-3221.0) cont'd

<u>From</u>	<u>To</u>	<u>Fractures</u>
3194.7	3195.5	Vertical, parted
3195.7	3201.9	Zone of closely-spaced, small inclined fractures
3212.0	3217.0	Zone of closely-spaced, small inclined fractures
3217.5	3217.5	Slickensided 60° surface

<u>From</u>	<u>To</u>	<u>Other Features</u>
3175.3	3175.4	Very calcareous zone
3181.3	3181.5	Very calcareous zone
3182.9	3183.0	Very calcareous zone
3183.7	3183.8	Very calcareous zone
3188.4	3188.6	Very calcareous zone
3193.1	3193.9	Very calcareous zone
3204.7	3204.7	Very calcareous layer
3207.3	3207.3	Very calcareous layer
3207.8	3207.8	Very calcareous layer
3211.0	3211.4	Calcareous zone
3211.8	3212.0	Calcareous zone
3213.3	3213.5	Very calcareous zone
3205.9	3205.9	Pyrite nodule
3218.0	3218.0	Pyrite nodule

Run #15 (3221-3280.5)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3221.0	3225.5	Medium gray-green shale with many dark gray bands
3225.5	3225.7	Very dark gray shale
3225.7	3225.8	Medium gray-green shale
3225.8	3226.7	Very dark gray shale
3226.7	3229.3	Medium gray-green shale
3229.3	3229.7	Dark gray shale
3229.7	3230.9	Medium gray-green shale
3230.9	3231.2	Light gray-green shale
3231.2	3233.1	Dark gray and medium dark gray banded shale
3233.1	3238.2	Medium gray-green shale
3238.2	3238.7	Very dark gray shale
3238.7	3239.8	Medium gray-green shale
3239.8	3239.9	Light gray shale
3239.9	3244.1	Medium gray-green shale
3244.1	3244.7	Dark gray and medium gray-green banded shale
3244.7	3248.2	Medium gray-green shale
3248.2	3249.4	Very dark gray shale
3249.4	3251.0	Medium gray green shale
3251.0	3252.0	Medium gray-green and dark gray banded shale
3252.0	3255.0	Medium gray-green shale
3255.0	3256.0	Medium gray-green and dark gray banded
3256.0	3261.0	Medium gray-green shale
3261.0	3261.6	Dark gray shale
3261.6	3265.8	Medium gray-green shale
3265.8	3266.5	Dark gray shale
3266.5	3270.3	Medium gray green shale
3270.3	3271.3	Medium gray-green and dark gray banded
3271.3	3271.8	Dark gray shale
3271.8	3273.3	Medium gray-green shale
3273.3	3274.9	Medium gray-green and dark gray banded shale
3274.9	3276.3	Dark gray
3276.3	3276.9	Medium green-gray shale
3276.9	3280.5	Medium dark gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3221.0	3226.4	Small, closely spaced fractures
3226.7	3228.1	Vertical, parted
3229.3	3230.1	Vertical, parted
3230.0	3230.8	Vertical, parted
3231.4	3234.5	Small, closely spaced fractures
3235.5	3240.3	Vertical, parted
3243.2	3244.2	Vertical, parted
3247.5	3248.2	Vertical, parted
3248.7	3249.8	Vertical, parted
3250.8	3251.7	Vertical, parted
3255.0	3256.0	Vertical, parted
3260.0	3261.6	Vertical, parted
3264.8	3265.5	Vertical, parted

Run #15 (3221-3280.5) cont'd

<u>From</u>	<u>To</u>	
3265.8	3267.0	Vertical, parted
3278.4	3279.6	Vertical, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
3230.9	3231.2	Very calcareous layer
3239.8	3239.9	Very calcareous layer
3267.4	3267.4	Very calcareous layer
3267.8	3267.8	Very calcareous layer
3278.3	3278.3	Very calcareous layer
3279.4	3279.4	Very calcareous layer

Run #16 (3280.5-3318.5)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3280.5	3281.2	Medium gray-green shale
3281.2	3281.5	Very dark gray shale
3281.5	3283.0	Medium gray-green shale
3283.0	3283.9	Medium gray-green and dark gray laminated shale
3283.9	3284.8	Medium gray-green shale
3284.8	3285.3	Very dark gray and medium gray-green laminated shale
3285.3	3285.7	Very dark gray shale
3285.7	3285.8	Medium gray-green shale
3285.8	3286.0	Very dark gray shale
3286.0	3287.2	Medium gray-green shale
3287/2	3288.1	Very dark gray shale
3288.1	3288.9	Medium gray-green and very dark gray laminated shale
3288.9	3289.9	Very dark gray shale
3289.9	3290.9	Medium gray-green shale
3290.9	3291.5	Very dark gray shale
3291.5	3294.9	Medium gray-green shale
3294.9	3295.3	Very dark gray shale
3295.3	3297.1	Medium gray-green shale
3297.1	3297.2	Black shale
3297.2	3298.5	Medium gray-green shale
3298.5	3299.6	Dark gray-green shale
3299.6	3300.1	Very dark gray shale
3300.1	3301.7	Dark gray-green shale
3301.7	3302.0	Very dark gray shale
3302.0	3302.8	Medium gray-green shale
3302.8	3304.6	Dark gray-green shale
3304.6	3304.9	Very dark gray shale
3304.9	3305.5	Medium gray-green shale
3305.5	3308.2	Dark gray-green shale
3308.2	3308.4	Very dark gray shale
3308.4	3309.0	Dark gray-green shale
3309.0	3309.3	Very dark gray shale
3309.3	3313.0	Dark gray-green shale
3313.0	3313.4	Very dark gray shale
3313.4	3315.0	Dark gray-green shale
3315.0	3315.2	Very dark gray shale
3315.2	3315.9	Medium gray shale
3315.9	3316.4	Very dark gray shale
3316.4	3317.0	Medium gray shale
3317.0	3317.4	Very dark gray and black shale
3317.4	3320.0	Medium gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3304.3	3305.9	Vertical, parted



Run #17 (3320.0-3379.1)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3320.0	3323.1	Medium gray-green shale
3323.1	3324.4	Dark gray shale
3324.4	3327.8	Medium gray-green shale
3327.8	3329.3	Dark gray shale
3329.3	3330.7	Medium gray-green shale
3330.7	3330.9	Dark gray shale
3330.9	3334.7	Medium gray shale
3334.7	3335.1	Dark gray shale
3335.1	3337.5	Medium gray shale
3337.5	3339.5	Dark gray shale
3339.5	3340.5	Medium gray shale
3340.5	3344.2	Medium gray-green shale
3344.2	3344.3	Light gray argillaceous limestone
3344.3	3344.4	Dark gray-green shale
3344.4	3346.5	Dark gray shale
3346.5	3348.5	Dark gray-green shale
3348.5	3349.0	Medium gray-green shale
3349.0	3350.0	Medium gray shale
3350.0	3350.4	Dark gray-green shale
3350.4	3351.6	Medium and dark gray-green shale
3351.6	3351.7	Medium gray shale
3351.7	3354.5	Medium gray-green and dark gray laminated shale
3354.5	3357.2	Very dark gray and dark gray-green laminated shale
3357.2	3357.3	Medium gray shale
3357.3	3360.9	Very dark gray shale
3360.9	3363.2	Very dark gray to black shale
3363.2	3364.5	Very dark gray shale
3364.5	3365.1	Black shale
3365.1	3366.5	Very dark gray shale
3366.5	3367.0	Black shale
3367.0	3368.8	Very dark gray shale
3368.8	3369.8	Dark gray-green shale
3369.8	3372.7	Very dark gray to black shale
3372.7	3373.2	Very dark gray and medium gray laminated shale
3373.2	3379.1	Very dark gray to black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
3320.5	3320.5	Slickensided horizontal surface
3322.7	3323.3	Vertical, parted
3323.4	3324.3	Vertical, parted
3324.3	3325.2	Vertical, parted
3326.7	3327.8	Vertical, parted
3330.5	3339.0	Vertical, parted
3339.4	3342.5	Vertical, parted

Run #18 (3379.1-3420.4)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
3379.1	3390.9	Very dark gray to black shale
3390.9	3391.6	Medium gray shale
3391.6	3392.1	Black shale
3392.1	3393.5	Very dark gray shale
3393.5	3393.6	Medium gray-green shale
3393.6	3393.7	Medium gray shale
3393.7	3397.6	Very dark gray shale
3397.6	3397.9	Dark gray-green shale
3397.9	3399.2	Very dark gray to black shale
3399.2	3399.3	Very dark gray shale
3399.3	3403.6	Black shale
3403.6	3404.6	Very dark gray to black argillaceous limestone
3404.6	3405.1	Medium gray argillaceous limestone
3405.1	3407.1	Very dark gray to black shale
3407.1	3409.0	Brownish black shale
3409.0	3414.5	Very dark gray to black shale
3414.5	3420.4	Light to medium gray limestone

<u>From</u>	<u>To</u>	<u>Fractures</u>
3380.1	3382.3	Vertical, parted
3383.4	3385.1	Vertical, parted
3385.1	3387.1	Vertical, parted
3387.1	3387.6	Inclined slickensided surface
3388.5	3389.1	Three intersecting slickensided surfaces
3389.5	3390.0	Inclined slickensided surface
3389.6	3390.8	Vertical, parted
3390.2	3391.6	Intersecting slickensided surfaces
3391.6	3394.1	Vertical, parted
3395.7	3396.9	Vertical, parted
3401.0	3401.4	Inclined slickensided surface
3403.1	3403.9	Vertical, parted
3413.4	3413.4	Horizontal slickensided surface
3414.1	3414.1	Horizontal slickensided surface

<u>From</u>	<u>To</u>	<u>Other Features</u>
3399.3	3399.4	Very calcareous zone with calcite stringers
3403.6	3405.1	Very calcareous zone
3414.5	3420.4	Very calcareous zone
3390.9	3390.9	Pyrite lamina
3397.7	3399.7	Pyrite lamina
3499.8	3400.8	Pyrite blebs
3403.1	3403.1	Pyrite lamina
3408.2	3408.5	Pyrite laminae and blebs

Description of Cored Shale Intervals  
from Well Number 20,338, Wise Co., Va.

Run #1 (4870'-4928.6)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
4870	4928.6	Entire run represented by black shale
<u>From</u>	<u>To</u>	<u>Fractures</u>
4885.2	4885.7	Slickensided, near vertical, parted
4894.2	4894.2	Slickensided horizontal plane
4896.2	4897.6	70° inclined, mineralized
4905.7	4905.7	Slickensided, horizontal plane
4910.5	4910.5	Slickensided, horizontal plane
4914.0	4914.5	3 vertical mineralized fractures, horizontal slickensided surface
4918.0	4918.0	3 near horizontal mineralized fractures, with a slickensided surface

<u>From</u>	<u>To</u>	<u>Other Features</u>
4871.5	4871.7	3 anhydrite-carbonate layers
4872.3	4872.3	Anhydrite-carbonate layer, with 2 pyrite blebs
4872.5	4873.3	3 anhydrite-carbonate layers
4873.5	4874.2	Numerous pyrite blebs
4875.4	4875.6	3 anhydrite-carbonate layers
4876.9	4877.1	3 anhydrite-carbonate layers
4877.8	4877.8	Anhydrite-carbonate layer
4878.2	4878.5	Numerous anhydrite-carbonate layers
4879.4	4879.4	2 anhydrite-carbonate layers
4882.2	4882.2	2 anhydrite-carbonate layers
4882.6	4882.7	2 anhydrite-carbonate layers
4884.2	4884.3	2 anhydrite-carbonate layers
4884.8	4885.7	Several anhydrite-carbonate layers
4888.8	4888.8	Anhydrite-carbonate layer
4891.4	4891.4	Anhydrite-carbonate layer
4892.6	4892.6	Anhydrite-carbonate layer
4892.9	4892.9	Anhydrite-carbonate layer
4893.0	4893.0	Anhydrite-carbonate layer
4893.7	4893.7	Pyrite filled layer
4894.6	4894.8	Numerous anhydrite-carbonate layers
4897.0	4897.2	Numerous anhydrite-carbonate layers
4900.0	4900.0	Anhydrite-carbonate layer
4900.4	4900.4	Anhydrite-carbonate layer
4901.3	4901.3	Anhydrite-carbonate layer
4901.5	4901.5	Anhydrite-carbonate layer
4903.1	4903.1	Anhydrite-carbonate layer with large pyrite bleb
4903.3	4904.3	4 anhydrite-carbonate layers
4904.5	4904.5	Anhydrite-carbonate layer with large pyrite bleb
4904.7	4904.7	Anhydrite-carbonate layer
4905.0	4905.0	Anhydrite-carbonate layer with 2 pyrite blebs
4905.5	4905.9	3 anhydrite-carbonate layers
4907.0	4907.2	Several anhydrite-carbonate layers

Run #1 (Cont.) (4908.7-4928.0)

<u>From</u>	<u>To</u>	<u>Other Features</u>
4908.7	4908.8	Anhydrite-carbonate layer with apparent fossil fragments
4911.0	4911.0	2 pyrite layers
4911.2	4911.5	2 anhydrite layers with 1 pyrite layer
4914.6	4914.6	Anhydrite-carbonate and pyrite layer
4919.0	4919.2	3 anhydrite-carbonate layers
4919.3	4919.3	Large pyrite bleb
4919.4	4928.0	Numerous anhydrite-carbonate layers

Gas bleeding throughout entire core

Run #2 (4928.6 - 4985.5)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
4928.6	4943.6	Black shale
4943.6	4945.1	Black shale with dark gray laminae
4945.1	4945.9	Medium dark gray shale
4945.9	4946.9	Medium dark gray and dark gray shale, banded and laminated
4946.9	4948.1	Dark gray shale with medium gray silty shale laminae
4948.1	4948.3	Medium dark gray shale
4948.3	4948.5	Medium dark gray shale with medium gray silty shale laminated in bedding or loading feature
4948.5	4950.2	Medium dark gray shale and dark gray bands laminae
4950.2	4951.1	Medium dark gray and black laminated shale with numerous silt laminae and loading feature pyrite blebs
4951.1	4951.8	Medium gray shale with medium dark to dark gray bands, loading features
4951.8	4955.3	Medium gray shale with dark gray and medium dark gray bands, silt laminae
4955.3	4955.8	Dark gray shale with medium dark gray laminae, with numerous siltstone
4955.8	4956.6	Medium gray shale with dark gray band
4956.6	4956.9	Medium gray silt shale
4956.9	4957.7	Medium gray shale with 2.5' dark and medium gray shale and silty laminae
4957.7	4958.5	Medium gray shale to black shale, silty laminae showing loading features
4958.5	4958.9	Medium gray shale
4958.9	4960.7	Medium dark gray shale with medium gray black shale, light gray laminae
4960.7	4961.9	Medium gray shale with medium dark gray laminae and some light gray laminae
4961.9	4962.5	Medium dark gray shale
4962.5	4963.4	Medium gray with occasional medium dark gray laminae
4963.4	4963.6	Light gray shale with slightly silty band
4963.6	4963.9	Medium gray shale
4963.9	4964.4	Light gray to medium gray shale with light gray silty laminae
4964.4	4964.9	Medium gray shale with a .15' light gray shale band
4964.9	4965.3	Light gray shale and medium gray shale cross bedded.
4965.3	4969.7	Medium gray shale with occasional light gray shale laminae
4969.7	4970.1	Medium gray shale
4970.1	4971.3	Light gray silty shale with medium dark gray laminae crossbedded
4971.3	4971.8	Medium gray shale
4971.8	4972.4	Light gray silty shale with medium gray laminae crossbedding
4972.4	4973.0	Medium gray shale with light gray laminated bands
4973.0	4980.8	Medium gray shale with sparse medium dark gray shale laminae and some light gray laminae
4980.8	4982.0	Medium gray with numerous medium dark gray bands, some light gray silty laminae with loading features
4982.0	4982.5	Medium gray shale

Run #2 (Cont.) (4982.5 - 4985.5)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
4982.5	4982.9	Black and medium gray shale laminae with light gray shale laminae
4982.9	4984.4	Medium gray shale with two .05 black shale bands
4984.4	4985.5	Medium dark gray and black laminated shale with light gray shale laminae

<u>From</u>	<u>To</u>	<u>Fractures</u>
4930.0	4930	Slickensided, horizontal plane, parted
4931.5	4931.5	Slickensided, horizontal plane, parted
4934.4	4934.4	Slickensided, horizontal plane, parted
4934.8	4934.8	Slickensided, horizontal plane, parted
4934.8	4935.2	4 near vertical, parted
4935.2	4935.2	Slickensided, horizontal plane, parted
4935.4	4935.4	Horizontal plane, mineralized
4935.7	4935.7	Slickensided, horizontal plane, parted
4937.7	4937.7	Slickensided, horizontal plane, parted
4939.5	4939.5	Slickensided, horizontal plane, parted
4940.5	4940.5	Slickensided, horizontal plane, parted
4941.7	4941.7	Slickensided, horizontal plane, parted
4942.2	4942.7	3 slickensided, horizontal planes, parted
4943.8	4945.0	3 slickensided, horizontal planes, parted
4946.3	4947.7	3 slickensided, horizontal planes, parted
4950.6	4950.6	Slickensided, horizontal plane, parted
4961.5	4961.5	Slickensided, horizontal plane, parted
4972.6	4972.6	Slickensided, horizontal plane, parted
4973.0	4973.7	2 60° inclined, slickensided, parted
4974.3	4974.3	Slickensided, horizontal plane, parted
4977.0	4977.4	4 slickensided, horizontal planes, parted
4978.4	4978.4	Slickensided, horizontal plane, parted
4980.5	4980.5	Slickensided, horizontal plane, parted
4981.7	4981.7	Slickensided, horizontal plane, parted
4984.2	4985.5	3 slickensided, horizontal planes, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
4929.0	4929.0	2 anhydrite - carbonate layers
4929.8	4930.0	3 anhydrite - carbonate layers
4931.2	4932.0	7 anhydrite - carbonate layers
4932.8	4932.8	Pyrite filled layer
4933.2	4935.3	6 anhydrite - carbonate layers
4937.2	4937.2	Large pyrite bleb
4938.5	4939.5	Numerous anhydrite - carbonate layers
4940.0	4940.3	3 anhydrite - carbonate layers
4943.7	4944.9	4 anhydrite - carbonate layers
4949.0	4950.6	5 pyrite blebs
4952.2	4955.0	Several pyrite blebs and layers
4960.2	4960.5	Large pyrite bleb with several filled layers

Run #2 (Cont.) (4962.2 - 4982.7)

<u>From</u>	<u>To</u>	<u>Other Features</u>
4962.2	4962.4	Several pyrite blebs and filled layers
4964.0	4964.0	2 pyrite filled layers
4965.0	4965.4	2 slightly anhydrite - carbonate layers
4982.7	4982.7	2 pyrite blebs

Slickensides in all directions but mostly horizontal  
for entire length of core.

Run #3 (5210-5220)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
5210	5220	Entire core cross-bedded medium dark gray shale with black shale bands and some silty laminae

<u>From</u>	<u>To</u>	<u>Fractures</u>
5211.4	5215.0	Slickensided, vertical, parted
5212.6	5212.6	Slickensided, horizontal plane, parted
5212.6	5214.2	Slickensided, near vertical, parted
5213.0	5213.0	Slickensided, horizontal plane, parted
5215.0	5215.8	50° inclined, slickensided, parted
5216.0	5217.7	70° inclined, slickensided, parted
5218.0	5218.0	Slickensided plane, parted
5219.1	5219.8	80° inclined, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
5210.0	5210.8	Numerous anhydrite - carbonate layers
5213.6	5214.0	5 anhydrite - carbonate layers
5215.1	5217.6	5 anhydrite - carbonate layers
5218.6	5218.8	3 anhydrite - carbonate layers



Run #4 (5220-5251)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
5220.0	5236.0	Medium dark gray shale with some black shale bands and laminae. Occasional silty streaks
5236.0	5236.3	Light gray silty zone
5236.3	5237.0	Medium dark gray shale with black shale bands
5237.0	5237.3	Light gray silty zone
5237.3	5249.2	Medium dark gray shale with black shale bands
5249.2	5251.0	Primarily black shale with some black shale bands

<u>From</u>	<u>To</u>	<u>Fractures</u>
5222.0	5223.0	Vertical, parted
5222.4	5222.4	Slickensided, horizontal plane, parted
5223.8	5224.4	45° inclined, slickensided, parted
5245.6	5246.4	Near vertical, parted
5246.2	5246.2	Slickensided, horizontal plane, parted
5247.0	5248.5	Vertical, parted
5249.2	5250.2	Vertical, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
5247.2	5247.2	Pyrite bleb
5247.7	5247.7	Pyrite bleb
5248.4	5248.4	Pyrite bleb
5248.9	5248.9	Pyrite bleb

Run #5 (5251-5310)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
5251.0	5255.4	Black shale with medium dark gray and dark gray shale laminae and bands, numerous anhydrite stringers
5255.4	5258.2	Dark gray shale and black shale layers with some worm burrows
5258.2	5260.7	Black shale and dark gray laminated shale
5260.7	5264.4	Medium dark gray shale with black shale bands and laminae, pyritized worm burrows near top
5264.4	5267.8	Black shale with medium dark gray shale bands and laminae
5267.8	5268.7	Black shale
5268.7	5272.2	Black shale with medium dark gray shale bands and laminae
5272.2	5273.4	Medium dark gray shale
5273.4	5278.9	Black shale with medium dark gray shale bands and laminae
5278.9	5280.5	Mostly medium dark gray shale with some dark gray and black shale bands and laminae
5280.5	5286.9	Black shale with dark gray shale bands and laminae and a few anhydrite seams
5286.9	5287.7	Medium dark gray shale with one .1' black shale band
5287.7	5292.3	Black shale with dark gray shale bands and laminae
5292.3	5292.7	Medium dark gray shale
5292.7	5310.0	Black shale with dark gray shale laminae and bands

<u>From</u>	<u>To</u>	<u>Fractures</u>
5253.3	5253.3	Slickensided, horizontal plane, parted
5263.5	5263.5	Slickensided, horizontal plane, parted
5266.4	5266.4	Slickensided, horizontal plane, parted
5268.8	5268.8	Slickensided, horizontal plane, parted
5270.0	5270.0	Slickensided, horizontal plane, parted
5274.6	5274.6	Slickensided, horizontal plane, parted
5275.2	5275.2	Slickensided, horizontal plane, parted
5285.2	5286.0	Near vertical to 70° incline
5292.1	5292.1	Slickensided, horizontal plane, parted
5305.0	5307.0	Vertical, irregular surface, parted
5307.5	5308.4	Vertical, irregular surface, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
5251.0	5254.8	Numerous anhydrite stringers
5258.3	5258.5	3 anhydrite - carbonate layers
5264.4	5264.4	Anhydrite - carbonate layer
5265.2	5266.0	5 anhydrite - carbonate layers
5267.3	5267.3	Anhydrite - carbonate layer
5268.5	5268.5	Anhydrite - carbonate layer
5271.7	5271.7	Anhydrite - carbonate layer
5272.8	5272.8	Anhydrite - carbonate layer

Run #5 (Cont.) 5274.7 - 5307.8

<u>From</u>	<u>To</u>	<u>Other Features</u>
5274.7	5274.7	Anhydrite - carbonate layer
5276.0	5277.0	Numerous anhydrite - carbonate layers
5277.3	5277.3	Anhydrite - carbonate layer
5278.0	5278.0	Anhydrite - carbonate layer
5278.4	5278.4	Anhydrite - carbonate layer
5279.0	5279.2	2 anhydrite - carbonate layers
5282.8	5282.8	Anhydrite - carbonate layer
5284.7	5284.7	Anhydrite - carbonate layer
5285.0	5285.2	2 anhydrite - carbonate layers
5285.4	5285.4	2 anhydrite - carbonate layers
5292.8	5292.8	Anhydrite - carbonate layer
5299.1	5299.4	2 anhydrite - carbonate layers
5299.8	5299.9	Several thin anhydrite - carbonate layers
5301.6	5301.6	Anhydrite - carbonate layer
5303.8	5303.8	Anhydrite - carbonate layer
5305.0	5305.0	Anhydrite - carbonate layer
5307.8	5307.8	Anhydrite - carbonate layer

Gas leaking along most of core.

Run #6 (5310-5345)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
5310.0	5313.8	Black shale
5313.8	5314.0	Medium dark gray shale
5314.0	5314.3	Black shale
5314.3	5314.5	Medium dark gray shale
5314.5	5318.2	Black shale
5318.2	5318.5	Medium dark gray shale
5318.5	5325.7	Black shale
5325.7	5327.7	Medium dark gray shale with black shale bands
5327.7	5329.0	Black shale with one band of medium dark to light gray shale .2' long
5329.0	5332.9	Medium dark gray shale with 2 black shale bands. One calcareous band at top and a few pyrite worm burrows
5332.9	5335.0	Black shale
5335.0	5337.5	Medium dark gray shale with 3 black shale bands .1' to .2' in width
5337.5	5339.0	Black shale
5339.0	5339.5	Medium dark gray shale
5339.5	5341.1	Black shale
5341.1	5341.3	Medium dark gray band
5341.3	5345.0	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
5311.5	5312.6	9. slickensided, horizontal plane
5314.2	5314.2	Horizontal plane
5319.2	5319.2	Slickensided, horizontal plane, parted
5320.0	5320.0	Slickensided, horizontal plane, parted
5321.3	5321.3	2 slickensided, horizontal planes, parted
5322.4	5322.4	Slickensided, horizontal plane, parted
5334.2	5334.2	Slickensided, horizontal plane, parted
5336.0	5336.0	Slickensided, horizontal plane, parted
5336.9	5337.4	3 slickensided, horizontal planes, parted
5338.0	5338.4	4 slickensided, horizontal planes, parted
5339.6	5339.6	Slickensided, horizontal plane, parted
5340.3	5340.3	Slickensided, horizontal plane, parted
5341.3	5341.3	Slickensided, horizontal plane, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
5310.8	5310.8	Anhydrite - carbonate layer
5311.5	5312.4	4 anhydrite - carbonate layers
5313.0	5313.0	Anhydrite - carbonate layer
5318.2	5318.2	Anhydrite - carbonate layer
5319.9	5319.9	Anhydrite - carbonate layer
5321.5	5321.5	Anhydrite - carbonate layer
5322.8	5322.8	Anhydrite - carbonate layer
5329.9	5329.9	Pyrite bleb
5330.3	5330.3	Pyrite bleb
5330.8	5330.8	Pyrite bleb

Run #6 (Cont'd) (5331.4-5337.0)

<u>From</u>	<u>To</u>	<u>Other Features</u>
5331.4	5331.4	Pyrite bleb
5332.4	5332.4	Pyrite bleb
5335.6	5335.6	Pyrite bleb
5336.1	5336.1	Pyrite bleb
5337.0	5337.0	Pyrite bleb

Considerable gas bleeding along core.

Run #7 (5345-5360)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
5345.0	5345.9	Black shale
5345.9	5347.9	Half black shale and half dark gray shale
5347.9	5351.0	Black shale
5351.0	5351.3	Medium dark gray shale
5351.3	5354.0	Black shale
5354.0	5354.5	Medium dark gray shale
5354.5	5354.9	Black shale
5354.9	5355.2	Medium dark gray shale
5355.2	5357.3	Black shale
5357.3	5357.5	Medium dark gray shale
5357.5	5360.0	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
5352.2	5352.3	2 slickensided, horizontal planes, parted
5352.5	5352.8	3 slickensided, horizontal planes, parted
5354.0	5354.0	Slickensided, horizontal plane, parted
5354.5	5354.5	Slickensided, horizontal plane, parted
5354.7	5354.7	Slickensided, horizontal plane, parted
5355.0	5355.8	7 slickensided, horizontal planes, parted
5356.2	5356.2	Slickensided, horizontal plane, parted
5356.4	5356.4	Slickensided, horizontal plane, parted
5357.7	5357.7	Slickensided, horizontal plane, parted
5358.2	5358.2	Slickensided, horizontal plane, parted
5359.1	5360.0	9 slickensided, horizontal planes, parted

Slickensides at 20° from horizontal

<u>From</u>	<u>To</u>	<u>Other Features</u>
5346.9	5346.9	Anhydrite - carbonate layer
5351.4	5351.4	Anhydrite - carbonate layer
5352.7	5352.7	2 anhydrite - carbonate layers
5354.2	5354.2	Anhydrite - carbonate layer
5354.7	5354.7	Anhydrite - carbonate layer
5355.8	5355.8	Anhydrite - carbonate layer
5356.2	5356.2	2 anhydrite - carbonate layers

Run #8 (5360-5392)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
5360.0	5380.9	Black shale
5380.9	5382.3	Medium gray shale
5382.3	5383.1	Black shale
5383.1	5383.7	Medium dark gray shale
5383.7	5384.2	Black shale
5384.2	5384.6	Dark gray shale
5384.6	5385.3	Black shale
5385.3	5385.8	Dark gray shale
5385.8	5388.0	Black shale
5388.0	5388.2	Medium dark gray shale
5388.2	5389.6	Black shale
5389.6	5392.0	Medium gray shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
5360.4	5360.6	3 slickensided, horizontal planes, parted
5361.4	5361.4	Slickensided, horizontal plane, parted
5361.8	5361.8	Slickensided, horizontal plane, parted
5362.5	5362.5	Slickensided, horizontal plane, parted
5363.1	5363.1	Slickensided, horizontal plane, parted
5367.6	5367.6	Slickensided, horizontal plane, parted
5368.0	5368.0	Slickensided, horizontal plane, parted
5370.8	5370.8	Slickensided, horizontal plane, parted
5371.2	5371.2	Slickensided, horizontal plane, parted
5373.3	5373.8	3 Slickensided, horizontal planes, parted
5375.4	5375.6	2 slickensided, horizontal planes, parted
5376.0	5378.0	12 slickensided, horizontal planes, parted
5378.5	5378.5	Slickensided, horizontal plane, parted
5379.0	5379.0	Slickensided, horizontal plane, parted
5379.3	5379.3	Slickensided, horizontal plane, parted
5379.6	5379.6	Slickensided, horizontal plane, parted
5380.9	5380.9	Slickensided, horizontal plane, parted
5381.4	5381.4	Slickensided, horizontal plane, parted
5381.8	5381.8	Slickensided, horizontal plane, parted
5382.0	5382.0	Slickensided, horizontal plane, parted
5383.3	5383.3	Slickensided, horizontal plane, parted
5384.1	5384.1	Slickensided, horizontal plane, parted
5384.5	5384.5	Slickensided, horizontal plane, parted
5384.7	5384.7	Slickensided, horizontal plane, parted
5387.0	5387.0	Slickensided, horizontal plane, parted
5388.0	5388.0	Slickensided, horizontal plane, parted
5388.2	5388.2	Slickensided, horizontal plane, parted
5389.1	5390.1	6 slickensided, horizontal planes, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
5362.9	5362.9	Anhydrite - carbonate layer
5380.6	5380.6	Large pyrite bleb
5386.6	5386.6	Pyrite bleb

Run #8 (Cont'd) (5388.5-5389.6)

<u>From</u>	<u>To</u>	<u>Other Features</u>
5388.5	5388.5	Pyrite layer
5389.0	5389.0	Pyrite layer
5389.6	5389.6	Pyrite layer

Gas bleeding profusely through entire core.



Run #9 (5392-5455)

<u>From</u>	<u>To</u>	<u>Lithologic Description</u>
5392.0	5393.0	Medium gray shale with numerous 45° filled hair-line fractures
5393.0	5397.6	Medium dark gray shale with worm burrows near base
5397.6	5399.3	Black shale
5399.3	5399.6	Medium dark gray shale and black shale with worm burrows
5399.6	5402.5	Black shale with some very dark gray shale
5402.5	5405.0	Medium dark gray shale with some black shale bands and laminae
5405.0	5406.7	Black shale with medium dark gray shale bands
5406.7	5410.3	Medium dark gray shale with 3 black shale bands approximately .25' - .41' wide
5410.3	5411.6	Black shale with a few dark gray shale laminae
5411.6	5413.0	Medium dark gray shale with one .3' wide black shale band with some worm burrows
5413.0	5413.7	Black shale
5413.7	5414.1	Medium dark gray shale band
5414.1	5418.4	Black shale with olive black shale layers
5418.4	5420.5	Black shale with olive black shale bands
5420.5	5422.4	Olive black shale with dark gray shale and black shale bands
5422.4	5428.3	Black shale
5428.3	5429.8	Black shale with dark gray shale bands
5429.8	5430.7	Black shale
5430.7	5431.1	Black shale and dark gray shale
5431.1	5434.4	Black shale
5434.4	5435.0	Black shale and dark gray shale with worm burrows
5435.0	5437.7	Black shale
5437.7	5438.0	Dark gray shale and black shale with worm burrows
5438.0	5439.5	Black shale
5439.5	5440.2	Dark gray shale and black shale with worm burrows
5440.2	5445.0	Black shale

<u>From</u>	<u>To</u>	<u>Fractures</u>
5393.0	5395.5	18 slickensided fractures, at 70° parted
5396.0	5396.5	5 slickensided, horizontal planes, parted
5396.8	5396.8	Slickensided, horizontal plane, parted
5397.2	5397.5	3 slickensided, horizontal planes, parted
5397.8	5398.2	4 slickensided, horizontal planes, parted
5398.6	5398.6	Slickensided, horizontal plane, parted
5399.6	5399.6	Slickensided, horizontal plane, parted
5400.0	5400.6	Near vertical
5401.0	5402.0	Near vertical
5402.4	5402.4	Slickensided, horizontal plane, parted
5404.4	5404.4	Slickensided, horizontal plane, parted
5405.0	5405.2	3 slickensided, horizontal planes, parted
5410.2	5410.2	Slickensided, horizontal plane, parted

Run #9 (Cont'd) (5411.3-5444.4)

<u>From</u>	<u>To</u>	<u>Fractures</u>
5411.3	5411.3	Slickensided, horizontal plane, parted
5414.5	5414.5	Slickensided, horizontal plane, parted
5416.4	5416.4	Slickensided, horizontal plane, parted
5416.9	5416.9	Slickensided, horizontal plane, parted
5417.3	5417.3	Slickensided, horizontal plane, parted
5417.4	5418.0	Near vertical
5428.0	5428.9	Near vertical
5432.6	5432.8	2 slickensided, horizontal planes, parted
5433.9	5434.0	2 slickensided, horizontal planes, parted
5434.4	5434.4	Slickensided, horizontal plane, parted
5434.7	5434.7	Slickensided, horizontal plane, parted
5435.0	5435.1	2 slickensided, horizontal planes, parted
5435.3	5435.5	3 slickensided, horizontal planes, parted
5435.7	5435.7	Slickensided, horizontal plane, parted
5435.9	5435.9	Slickensided, horizontal plane, parted
5436.1	5436.5	5 slickensided, horizontal planes, parted
5436.7	5437.4	5 slickensided, horizontal planes, parted
5437.8	5437.8	Slickensided, horizontal plane, parted
5438.0	5438.0	Slickensided, horizontal plane, parted
5438.9	5438.9	Slickensided, horizontal plane, parted
5439.4	5439.4	Slickensided, horizontal plane, parted
5442.0	5442.0	Slickensided, horizontal plane, parted
5442.5	5442.5	Slickensided, horizontal plane, parted
5444.4	5444.4	Slickensided, horizontal plane, parted

<u>From</u>	<u>To</u>	<u>Other Features</u>
5398.0	5398.0	2 pyrite blebs
5399.0	5399.0	Large pyrite bleb
5400.0	5400.0	Pyrite bleb
5406.4	5406.4	Anhydrite - carbonate layer
5417.4	5417.5	Large pyrite bleb
5418.3	5418.3	Pyrite bleb
5419.0	5419.0	Large pyrite bleb
5420.0	5420.0	Distinct black kerogen layer
5420.1	5420.1	Pyrite bleb
5421.5	5421.5	Pyrite bleb
5423.4	5423.4	Pyrite bleb
5426.2	5426.2	Pyrite bleb
5428.3	5428.3	Pyrite bleb
5432.7	5432.7	Pyrite layer
5438.2	5438.2	2 pyrite blebs
5438.4	5438.4	Pyrite bleb

APPENDIX IV  
TERRA TEK REPORTS

IN SITU STRESS DETERMINATION IN THE DEVONIAN SHALE  
(IRA McCOY 20402) WITHIN THE ROME BASIN

By

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Submitted to

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TR 77-89  
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## RECOMMENDATIONS

1. The nature (direction, size and frequency) of the small-scale natural fracture system in the Devonian shale should be investigated. The intent in any MHF treatment is that the induced fracture will intersect the existing natural channel (fractures) available for gas flow. Hence, the proximity, at least at the wellbore, of the directions of both the natural fractures and the hydraulically induced one should be considered when a full-scale stimulation job is planned.

2. Difference in magnitudes of the minimum horizontal *in situ* stresses acting within the various layers in any stimulated well represent one of the most important parameters in fracture containment within the pay zone. Achievement of deeply penetrating fractures in the pay formation, which is associated with minimum excursion into the barrier layers, depends on controlling the BHTP in accordance with the difference in *in situ* stresses. Therefore, until better understanding of the factors affecting the *in situ* stress field is achieved, it is imperative to continue the present effort in measuring the *in situ* stresses in the pertinent formation prior to any MHF treatment.

### SUMMARY

The hydrofracturing technique has been used to determine the naturally occurring horizontal *in situ* stress field in the Upper Gray zone of the Devonian shales. The field experiment was performed at a depth of 2745 ft. (837m) in Ira McCoy #20402 well. The bearing of the induced fracture at the wellbore has been determined using an impression packer. Laboratory tests have been performed on core samples from the same formation to determine the rock resistance to fracturing.

Logging data and laboratory observations of core samples reveal the existence of small-scale (~0.25 - 0.5 in.) vertical cracks oriented at N 50° E to N 60° E. This direction corresponds with the direction of fracture propagation indicated by the impression packer. Furthermore, it appears to be aligned with the direction of the projection of the Rome Trough in the test region.

Analysis of all the available field and laboratory data along with the geologic considerations gives the following values and estimates of the *in situ* stress field:

Overburden Stress = 3210 psi (22.1 MPa)

Minimum Horizontal Stress = 2360 psi (16.3 MPa)

Maximum Horizontal Stress = 4390 psi (30.3 MPa) - estimated

Bearing of Induced Fracture = N 45° E to N 55° E

Hence, the induced fracture tends to align itself at the wellbore with the small-scale natural fracture system in that zone.

## ABSTRACT

The *in situ* stress field was determined at a depth of 837 m (2,745 feet) in Devonian Shale ("gray" shale) within the Rome Basin in West Virginia. Logging data and laboratory observations of core samples reveal vertical cracks oriented at N 50° - 60° E. Because of these cracks and their preferred orientation, a new approach based on fracture mechanics concepts is used to evaluate the *in situ* stresses from the field and laboratory data. The resulting prediction of the maximum horizontal stress ( $\sigma_{HMAX}$ ) is compared to that predicted by Haimson and Fairhurst's [1967] method; the latter method appears to overestimate the value of this stress component because the effect of loading the faces of any pre-existing crack is neglected.

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## INTRODUCTION

Recent attempts to stimulate natural gas production in low-permeability sandstones in the Western United States by massive hydraulic fracturing (MHF) have resulted in increases of flow of up to 8 times (J. Wroble, unpublished data 1976). However, many attempts have been unsuccessful, probably owing to certain unfavorable characteristics of the pay formations, such as permeability and deformation moduli [Randolph, 1976], all of which are affected by the magnitude of the *in situ* stress. Thus, knowledge of the stress state in the pay and surrounding formations is essential in the simulation of *in situ* conditions during laboratory experiments designed to measure the true subsurface characteristics. These measurements, in turn, make the analysis of phenomena such as containment of the fracture within the pay zone and *in situ* permeability more reliable. In addition, determination of the *in situ* state of stress at depth gives insight into what the breakdown pressure, extension pressure, and the direction of fracture will be during MHF. The difference in horizontal principal stresses is of particular interest because the direction of a hydraulic fracture will be determined by the prevailing direction of the minimum principal stress, provided that the difference in principal stresses is significantly greater than local fluctuations in principal stresses. Furthermore, knowledge of the direction of the minimum *in situ* principal stress and/or the preferred orientation of natural fracture systems is needed in deviated well technology (Komar, unpublished data, 1976a). In this technology the well bore is drilled

at a pre-determined inclination with its horizontal projection aligned with either the direction of minimum *in situ* principal stress or orthogonal to the direction of the natural fracture system.

Field techniques to determine the magnitudes and directions of *in situ* principal stresses include, among others, hydrofracturing. The method is a by-product of the hydraulic fracture stimulation technique. The conventional analysis of the results of hydrofracturing involves computation of the *in situ* stresses from the elasticity solution for a pressurized, smooth well bore in an isotropic, homogeneous elastic medium.

During a hydrofracturing experiment to determine *in situ* stresses in the Devonian Shales within the Rome Basin of West Virginia, it was observed that the natural fractures in the core samples from the test well (Ira McCoy 20402) violated the assumptions used in the conventional methods of calculating the *in situ* stress field [cf. Haimson and Fairhurst, 1967]. The problem was, therefore, approached using the principles of linear elastic fracture mechanics; and the results of this analysis suggest that the conventional analysis is incorrect in the determination of the maximum principal stress  $\sigma_{HMAX}$ . The error occurs because the mechanics of fracture initiation and fracture extension are ignored in the conventional method of calculating maximum compressive *in situ* stress from an elasticity solution for a pressurized cylindrical cavity.

The present report has been divided into two distinct sections. The first section deals with the conventional approach as applied to this particular experiment; a proposed new approach is presented in the second section. The differences between the two approaches for computing the

maximum *in situ* stress are examined and found to be due mainly to neglect of crack geometry and the loading of pre-existing cracks by the pressurized fluid in the former approach. From fracture mechanics principles, these factors have a marked effect on crack extension and should, therefore, be considered in the determination of  $\sigma_{HMAX}$ .

## THE OPEN-HOLE FIELD TESTS AND RESULTS

### Background

In hydrofracturing experiments designed to measure *in situ* stresses the hole is left open so that the orientation of the fracture can be determined after the formation has been broken down. The section of the hole to be tested is isolated by lowering "straddle packers" into position and then pressurizing the sealing components at each end of the device (see Figure 1). The "fracturing fluid" is then injected into the section between the upper and lower seals. Surface and, if possible, downhole recorders are used for continuously monitoring the fluid pressure. The pressure is raised slowly until the breakdown pressure ( $P_b$ ) is reached, i.e., the pressure at which the rock surrounding the hole fractures. If the flow rate remains constant after the breakdown pressure has been reached, the pressure will drop to a constant level, known as the extension pressure ( $P_e$ ) at which the fracture propagates. If the fluid flow is stopped, crack extension ceases and the pressure drops to an equilibrium value called the shut-in pressure ( $P_s$ ). Finally, to determine the orientation of the fracture, an impression packer is lowered into the test section and a trace is formed on the packer by extruding a soft rubber membrane into the fracture. A photograph of a downhole compass is taken and then correlated with a reference mark on the outside of the packer. For a more complete description, see Haimson [1968].

Applications of hydrofracturing are potentially unlimited in depth (except for unavailability of pumping equipment suitable for the necessary high pressure needed for great depths) and do not depend on the determination

of load-deformation response, as in the case of overcoring techniques. Since its inception, it has undergone theoretical development [cf. Hubbert and Willis; 1957, Scheidegger, 1960; Kehle, 1964; Haimson and Fairhurst, 1967] and has been applied in both laboratory experiments [Lamont and Jensen, 1963; Haimson and Fairhurst, 1967] and field experiments (cf. Raleigh, et al. [1972], Haimson, et al. [1973] and Bredenhoeft, et al. [1976]).

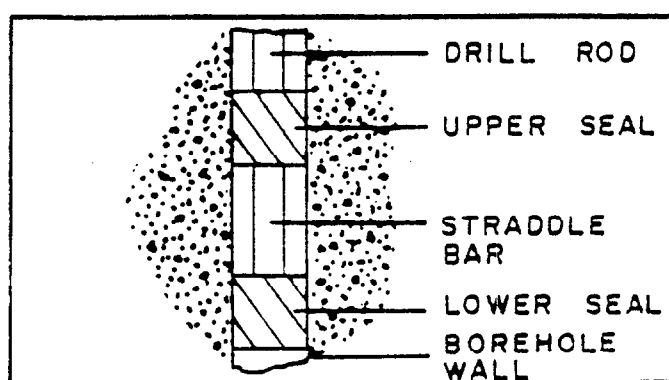


Figure 1. Configuration of a typical straddle packer

#### The Open-Hole Test

The open-hole test was conducted in the Upper Gray Zone of the Devonian Shales using a standard Lynes straddle packer for a 22.2cm (8-3/4 inch) hole. The straddle length was roughly 4.27m (14 ft.) and its center was located at a depth of 837m (2745 ft.). The packer and drill stem were then filled with a low-viscosity fluid (a combination of KCl, Macobar Drispak and water) that was used to drill the well. The packers were then pressurized to 10.5 MPa (1520 psi, downhole gage).

After the packer had been pressure-set, the formation was pressurized until the breakdown pressure was reached (see Figure 2). Both the pressure and flow rates were monitored at the surface and a downhole pressure monitor

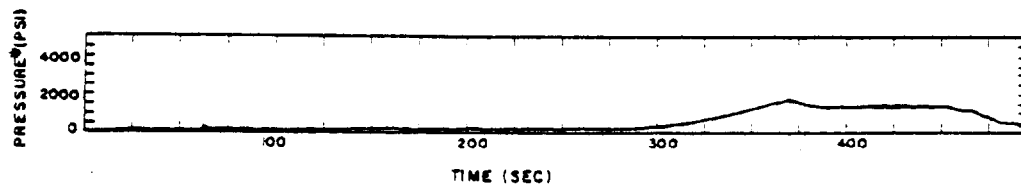


Figure 2. Test record indicating breakdown pressure and extension pressure

\* Add hydrostatic head equal to 8.4 MPa (1220 psi)

installed in the straddle bar confirmed the surface measurements. Figure 3 indicates the breakdown of the formation and the initial extension of the fracture in more detail. It was not possible to record the instantaneous shut-in pressure after breakdown because a pressure fitting at the top of the drill stem started leaking.

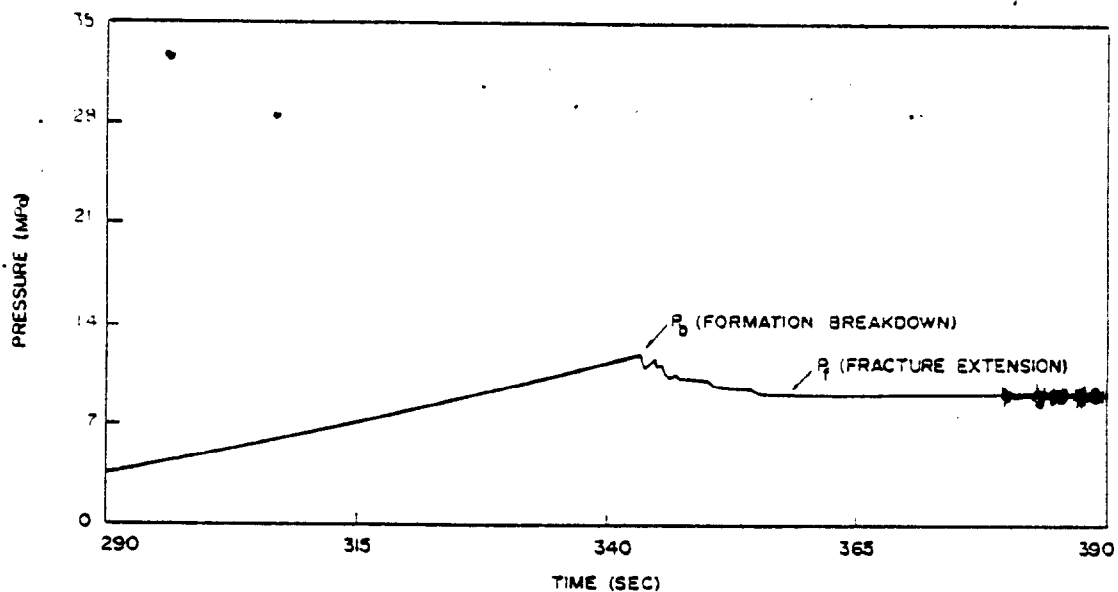


Figure 3. Test record indicating breakdown pressure and extension pressure.

\* Add hydrostatic head equal to 8.4 MPa (1220 psi)



The pressure fitting was replaced and the shut-in pressure was measured on the three successive runs, two of which are shown in Figures 4 and 5.

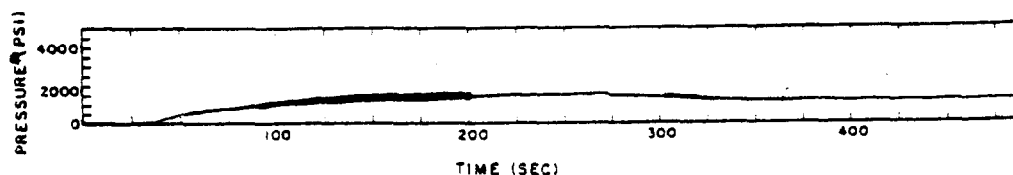


Figure 4. Fracture extension and shut-in Number 1.

\* Add hydrostatic head equal to 8.4 MPa (1220 psi)

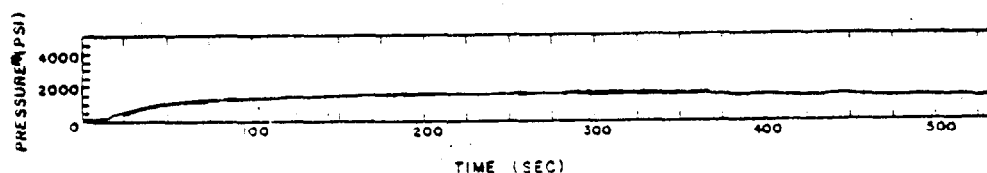


Figure 5. Fracture extension and shut-in Number 2.

\* Add hydrostatic head equal to 8.4 MPa (1220 psi)

The numerical results of the open-hole test are presented in Table I. Maximum variation between the successive measurements of the shut-in pressure was within  $\pm 6$  percent.

TABLE I

Results of the Open-Hole Test

Pressure Run	Breakdown Pressure ( $P_b$ ) Mpa (psi)	Shut-in Pressure ( $P_s$ ) Mpa (psi)
1	20.2 (2930)	---
2	---	16.4 (2380)
3	---	17.3 (2470)
4	---	15.3 (2215)

The straddle packer was removed from the hole and a Lynes impression packer was lowered to map the fracture that had been formed. The bottom of the impression was at 839m (2752 ft.); it extended 3.66m (12 ft.) with an 0.61m (2 ft.) blank in the middle. The impression packer indicated the formation of a very narrow (approximately 0.02 in.) vertical fracture over its entire length. The bearing of the fracture trace at the borehole surface was determined to be N 45° E to N 50° E using an Eastman Whipstock single-shot downhole camera.

#### Evaluation of the *In Situ* Stress Field

Determination of the complete *in situ* stress field can be achieved if the three principal stresses and their directions can be calculated. In general, several hydrofracturing experiments in non-coplanar boreholes are required to obtain the necessary information. This method was used to measure the *in situ* stress at the Nevada Test Site [Haimson, et. al. 1973]. If, however, the direction and magnitude of one of the principal stresses are known, the complexity of the experimental work needed to determine the other two principal stresses is reduced substantially. This situation is not uncommon since the vertical (overburden) stress is usually a principal stress and, except at very shallow depth, will generally exceed the minimum *in situ* principal stress.

If the rock is viewed as a homogeneous, isotropic, elastic medium with an isotropic failure criterion, it is easily shown that a crack propagating in this medium, due to fluid pressure acting on its face, will grow along the path of least resistance (i.e., it will extend in the plane perpendicular to the least compressive *in situ* principal stress). For a moderately long crack, the pressure required to hold the crack open, but

not extend it, will be slightly greater than the far-field stress acting normal to the fracture. Therefore, it follows that the shut-in pressure is approximately equal to the minimum horizontal compressive stress.

In this case, if the rock is assumed to fail when a critical tensile stress is reached, then the principal stress field can be calculated by applying Equations (1), (2) and (3).

$$\sigma_{HMAX} = T_o + 3P_s - P_b - P_o \quad (1)$$

$$\sigma_{OB} = \rho H \quad (2)$$

$$\sigma_{HMIN} = P_s \quad (3)$$

$\sigma_{HMAX}$  = total maximum horizontal compressive stress.

$\sigma_{HMIN}$  = total minimum horizontal compressive stress.

$\sigma_{OB}$  = total vertical principal stress due to overburden.

$\rho$  = specific weight of rock (kPa/m or Psi/ft)

$H$  = depth of the test zone (m or ft)

$P_o$  = formation pore pressure

$T_o$  = tensile strength of the rock

$P_s$  = shut-in pressure

$P_b$  = breakdown pressure

This solution, based on the elasticity solution for a pressurized cylindrical cavity in an infinite isotropic elastic continuum, was first proposed by Hubbert and Willis [1957] and Haimson and Fairhurst [1967]. It should be noted that Equations (1) and (3) are valid only for the case where the fracturing fluid does not penetrate the matrix of the formation material. Due to the low permeability of the Devonian Shales of the Rome Basin, neglect of fracturing-fluid penetration should not introduce major

errors into the analysis. Equations (1), (2) and (3) are not applicable if the vertical stress is less than the smallest horizontal stress, in which case a horizontal fracture will form and determination of the horizontal *in situ* principal stresses is not possible using conventional hydrofracturing techniques.

In addition to the magnitude of the *in situ* principal stresses obtained from the measured pressures and Equations (1) and (3), the impression packer provides the direction of the cracks that have formed at the borehole wall. If the material is isotropic, these cracks should be normal to the direction of the minimum principal stress.

#### Results of the Laboratory Experiments

Laboratory tests conducted in this study were used to supply the material properties necessary to obtain the *in situ* stresses from the field test. A series of six hollow-cylinder burst tests were conducted on oriented core samples recovered from the Gray Shale section at a depth of 823m (2699 ft.) to 845m (2770 ft.). These tests, when interpreted in terms of a critical tensile stress required for fracture, provided the tensile strength ( $T_0$ ) of the shale. A description of the experimental techniques used in these tests can be found elsewhere [Haimson and Fairhurst, 1967].

Briefly, the test consists of internal pressurization of a thick-walled cylindrical sample until failure occurs. The failure pressure,  $P_i$ , is measured and is related to the tensile strength of the rock,  $T_0$ , through the equation

$$T_0 = P_i \left( \frac{w^2 + 1}{w^2 - 1} \right) \quad (4)$$

where  $w$  is the ratio of the outer radius  $b$  to the inner radius  $a$  of the hollow cylinder. The tests reported here were carried out with  $a = 9.27\text{mm}$  (0.365 in.) and  $b = 96.5\text{mm}$  (3.80 in.).

In spite of the visual competence of the shale from the test section, nearly all of the core samples developed extensive horizontal fractures during transportation to the Terra Tek facility. The fractures are apparently due to bedding plane separation caused by the removal of the overburden stress. These fractures limited the length of the burst test specimens to less than 63mm (2.5 in). The results of these laboratory tests are shown in Table II.

TABLE II  
Results of Unjacketed Burst Tests

Sample Depth m (ft.)	$b/a = w$	$P_i$ Failure Pressure MPa (psi)	Remarks
823 (2699)	10.7	22.3 (3230)	Very competent sample, few fractures.
831 (2724)	10.7	7.7 (1120)	Failed on pre-existing fractures.
836 (2741)	10.7	11.3 (1640)	Failed on pre-existing fractures.
836.3 (2742)	10.7	2.3 (340)	Failed along a bedding-plane fracture.
842 (2761)	10.7	5.9 (850)	Failed along a bedding-plane fracture.
845 (2770)	10.7	19.2 (2790)	Very competent sample, very few fractures.

It is obvious that the results from the unjacketed burst tests can be grouped into two classes. Samples at 823m and 845m levels have an average tensile strength which is over twice as great as for samples at the 831m and 836m levels. Samples from depths of 823m and 845m had considerably fewer natural fractures than the samples from the 831m and 836m levels.

Both of the samples that failed at the lower pressures (831m and 836m) had groups of small, tight vertical fractures with the same directional trend, and the failure plane induced during the test was parallel to the pre-existing fractures. The orientation of pre-existing cracks was generally N 45-55° E, which agrees reasonably well with the bearing of the fracture trace at the well bore.

### Calculation of *In Situ* Stress Based on a Critical Tensile Strength Hypothesis

A comparison of the densities obtained from the Birdwell 3-D ultrasonic log and the density measurements done by Terra Tek indicates excellent agreement. Therefore, the log densities were averaged and used to calculate the overburden stress. The calculated value for the overburden stress  $\sigma_{OB}$  is 22.1 MPa (3210 psi).

The tensile strength used in the calculation of the maximum horizontal stress is taken from the results of theunjacketed burst of a core from the field test section that did not fail because of bedding plane fractures. The three measurements of the shut-in pressure were averaged to determine the value of the minimum horizontal stress,  $\sigma_{HMIN}$ . The pore pressure term in Equation (1) is estimated from a downhole-pressure-build-up test to be 1.7 MPa (250 psi) (Smith, unpublished data, 1976). A tabulation of the measured and calculated data at the mid-position of the fractured section (837m) is presented in Table III.

TABLE III

Calculation of *In Situ* Stress at 837m (2745 feet)  
for Isotropic Medium

Depth m(feet)	$P_o$ MPa(psi)	$P_p$ MPa(psi)	$P_s$ MPa(psi)	$T_{oavg}$ MPa(psi)	$P$ k Pa/m(psi/ft)	$\sigma_{OB} = \sigma_H$ MPa(psi)	$\sigma_{HMAX} = T_o + 3P_s - P_o - P_p$ MPa(psi)	$\sigma_{HMIN} = P_s$ MPa(psi)	Direction of $\sigma_{HMAX}$
837 (2745)	20.2 (2930)	1.7 (250)	16.3 (2360)	11.3 (1640)	26.4 (1.17)	22.1 (3210)	38.3 (5550)	16.3 (2360)	N45°-50°E

As a result of the scatter observed in the laboratory burst tests, the error bounds for the maximum horizontal stress are quite large. If, for example, the material in the test section between the packers were isotropic and homogeneous with a tensile strength of 21.1 MPa (3060 psi), corresponding to the average burst pressure of the very competent samples at 823m and 845m, then a maximum horizontal stress of 48.1 MPa (6980 psi) would be obtained. On the other hand, if the tensile strength of the rock were 7.7 MPa (1120 psi), corresponding to the minimum burst pressure when bedding plane fracture does not occur, then the maximum horizontal stress would be 34.7 MPa (5020 psi).

The wide range of measured burst pressures coupled with the effect of the known existing fractures observed in the recovered cores indicate that the shale medium has pronounced anisotropy with respect to its tensile strength. Figure 6 shows that the induced failure plane is parallel to the group of existing vertical fractures in the sample at the 831m (2726



Figure 6. Burst sample from 831m (2724 ft.)  
[ $P_i = 7.7$  MPa (1120 psi)]

ft.) level. Furthermore, examination of the fracture density logs from Ira McCoy well 20403 indicates a consistent set of vertical fractures through the entire cored section (see Figure 7). Azimuths of these fractures are within  $\pm 10$  degrees of the direction of the fracture created during the hydrofracturing test. Hence, the assumptions of a crack-free well bore and a critical-tensile-strength failure criterion do not appear to be appropriate.

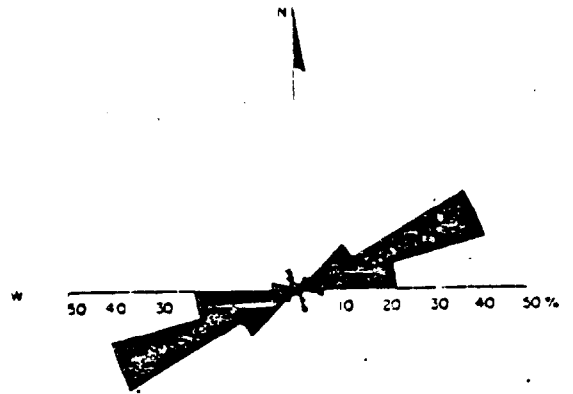


Figure 7. Histogram of the strikes of 605 vertical extension fractures mapped in Devonian Shale recovered from well No. 20403 (from Swolfs, *et al.*, 1976).

In order to account for the existence of the developed fracture system in the field test, the problem has been reanalyzed by applying the methods of linear elastic fracture mechanics. The effect of preferred crack orientation on fracture growth and the initiation of fracture are accounted for in the calculation of the *in situ* stress. The calculated stress state will be found to differ from the estimates obtained using the conventional formulation [Haimson and Fairhurst, 1967].



## FRACTURE MECHANICS ANALYSIS OF THE HYDROFRACTURING TEST

Growth of a crack inclined to the directions of the far-field *in situ* stresses and subjected to pressure on its faces can be analyzed by using fracture mechanics concepts in which linear elasticity is assumed and attention is given to the elevation of stresses near the crack tip.

While large stresses around the crack tip are usually accompanied by some plasticity, linear elastic fracture mechanics properly forms the basis for analyses when plastic deformation and other non-linear effects near the crack tip are confined to a small region within a linear elastic field.

The near-tip stress state in such a field is characterized by a single parameter. Irwin [1960] introduced the stress-intensity factor  $K$  as one such parameter--others include the J-integral [Rice, 1968] and the specific energy-release rate,  $G$  [Irwin, 1957]. These parameters, which are equivalent in linear elastic fracture mechanics, measure the intensity of the local stress field at the crack tip. They are determined by the applied loading, the crack configuration, and the geometry of the body. Cracks are expected to advance if the values of these parameters reach critical values characteristic of the material considered. On the other hand, if the loads acting on the body and its geometry are such that the value of  $K$ ,  $J$ , or  $G$  is less than this characteristic material property, then the crack is expected to remain stationary.

### Effect of Preferred Crack Orientation on Crack Growth in Hydraulic Fracturing

Consider a pressurized crack oriented at an arbitrary angle  $\alpha$  relative to the direction of minimum principal stress  $\sigma_3$  of the far-field stress system, illustrated in Figure 8. If the crack faces of length  $2L$  are subjected to a pressure  $p$ , the stress-intensity factors  $K_I$  (for the opening mode)

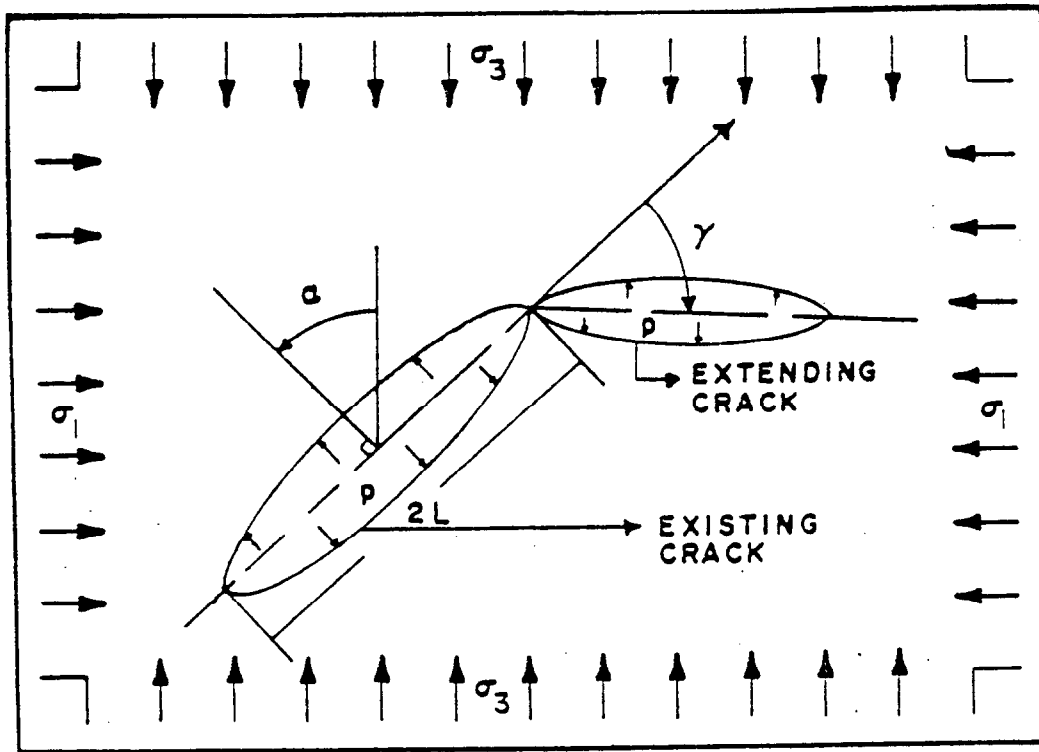


Figure 8. Skewed crack under far-field stress and internal pressure

and  $K_{II}$  (for in-plane shearing mode) for the existing crack are given by (cf. Rice, 1970, eq. 97):

$$K_I = \sqrt{\pi L} \{ p - \sigma_1 \sin^2 \alpha - \sigma_3 \cos^2 \alpha \} \quad (5)$$

$$K_{II} = \sqrt{\pi L} \{ \frac{1}{2} (\sigma_1 - \sigma_3) \sin^2 \alpha \} \quad (6)$$

where  $\sigma_1$  and  $\sigma_3$  denote, respectively, the maximum and minimum compressive far-field horizontal principal stresses.

If the existing crack extends in an arbitrary direction, as shown in Figure 8, then the energy-release rate  $\Gamma(\gamma)$  associated with extension in the direction  $\gamma$  will be given by the following equation [Hussain, et al., 1974]

$$\Gamma(\gamma) = \frac{4(1-\nu^2)}{E} \left( \frac{1}{3+\cos^2\gamma} \right) \left( \frac{\pi-\gamma}{\pi+\gamma} \right)^{\gamma/\pi} \left[ (1+3\cos^2\gamma)K_I^2 + 8\sin\gamma\cos\gamma K_I K_{II} + (9-5\cos^2\gamma)K_{II}^2 \right] \quad (7)$$

where  $K_I$  and  $K_{II}$  are given by Equations (5) and (6), and  $E$  and  $\nu$  denote the Young's modulus and Poisson's ratio of the material, respectively.

If the crack is assumed to advance in the direction,  $\gamma_{\max}$ , for which  $\Gamma(\gamma)$  is a maximum, then the relationship between the direction of crack advance,  $\gamma_{\max}$  and the ratio,  $(K_{II}/K_I)$ , is given by Figure 9 [Clifton, 1973; Palaniswamy and Knauss, 1972].

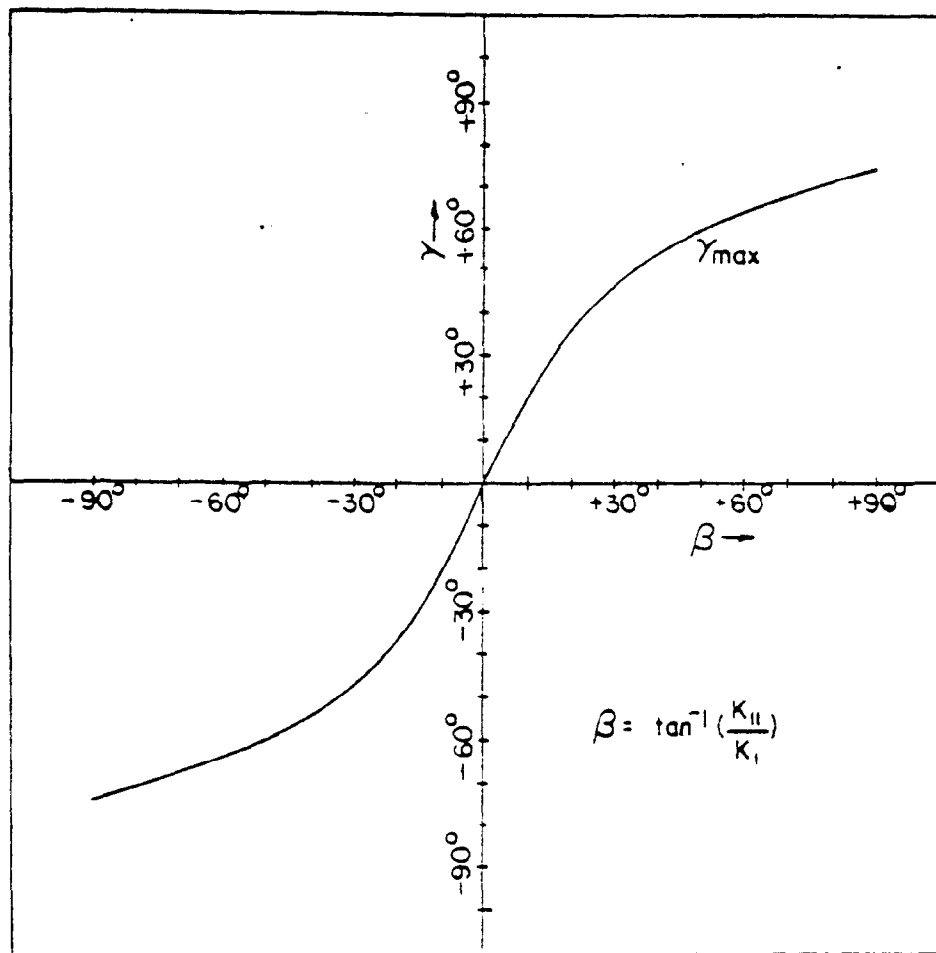


Figure 9. Orientation of propagating crack which maximizes energy-release rate (Clifton, 1974).

The energy release rate  $\Gamma(\gamma)$  is a positive definite function of  $K_I$  and  $K_{II}$ . Furthermore, substitution of (5) and (6) into (7) reveals that  $\Gamma(\gamma)$  is proportional to the crack length. Therefore, if cracks advance for a prescribed finite value of  $\Gamma(\gamma_{\max})$ , the only way a sufficiently long open crack cannot advance is for  $K_I$  and  $K_{II}$  to be equal to zero. From (5) and (6) the conditions  $K_I = 0$  and  $K_{II} = 0$  are satisfied for

$$p = \sigma_1 \sin^2 \alpha + \sigma_3 \cos^2 \alpha \quad (8a)$$

and

$$(\sigma_1 - \sigma_3) \sin 2\alpha = 0 \quad (8b)$$

In the former case,  $\alpha = 0$ , Equation (8a) requires that the pressure  $p$  equals the minimum horizontal compressive stress,  $\sigma_3$ , whereas, in the latter case,  $\alpha = \pi/2$ , the pressure  $p$  equals the maximum horizontal compressive stress  $\sigma_1$ . That is, sufficiently long open cracks can be stationary only if they are parallel to principal stress directions and if the pressure  $p$  is equal to the principal stress perpendicular to the crack face.

Whether the case  $\alpha = \pi/2$  or  $\alpha = 0$  is more likely to occur can be determined by noting whether an arbitrarily oriented crack tends to rotate to become perpendicular to  $\sigma_1$  or  $\sigma_3$ . From Figure 9, a crack tending to open due to internal pressure (i.e., a crack with  $K_I > 0$ ) will have a maximum energy release rate if it extends in the direction of positive  $\gamma$  for  $K_{II} > 0$  and negative  $\gamma$  for  $K_{II} < 0$ . Then, for  $(\sigma_1 - \sigma_3) > 0$ , equation (6) indicates that for  $\alpha$  in the interval  $(0, \pi/2)$  the predicted inclination  $\gamma$  of the extending crack is positive whereas  $\gamma$  is negative for  $\alpha$  in the interval  $(-\pi/2, 0)$ . In other words, the maximum energy release rate hypothesis predicts that the crack tends to extend in a direction which is more nearly perpendicular to the direction of minimum compressive stress than was the existing crack as

long as  $(\sigma_1 - \sigma_3)$  is not zero. This prediction is consistent with laboratory results [Brace and Bombolakis, 1963; Ingraffea and Heuze, 1976].

Thus, for long cracks the angle  $\alpha$  approaches zero so that, from (8a), the shut-in pressure,  $P_s$ , is given by

$$P_s = \sigma_3 = \sigma_{HMIN} \quad (9)$$

That is, for sufficiently long cracks the static pressure at which crack advance is imminent is independent of the crack length and the mechanical properties of the rock. The minimum crack length necessary for (9) to provide a good approximation of  $\sigma_{HMIN}$  can be estimated from (5) and an assumed critical value,  $K_{IC}$ , of the stress-intensity factor at which crack advance occurs. For  $K_{IC} = 40 \text{ MPa}(\text{mm})^{1/2}$  (See Table A1), the difference  $(P_s - \sigma_3)$  is less than 0.5 MPa (73 psi) for crack lengths greater than 2 meters. The injected volume of fluid in hydrofracturing experiments is sufficiently large that crack lengths are expected to be considerably greater than 2 meters. Hence, Equation (9) or (3), should allow reasonably accurate determination of the minimum *in situ* stress from the measured shut-in pressure.

#### Crack Initiation with a Pre-Existing Crack of Prescribed Orientation

An estimate of the maximum horizontal *in situ* stress,  $\sigma_{HMAX} = \sigma_1$ , can be obtained by considering the initial advance of an existing crack which intersects the bore hole at a prescribed orientation. A cylindrical hole with two radially opposed cracks subjected to an internal pressure  $p$  and the far-field stress state  $\sigma_1$  and  $\sigma_3$  is shown in Figure 10. The stress-intensity factor,  $K_I$ , at the crack tip for the case when the pressure  $p$  acts on the bore and the crack faces is given by [Johnson, *et. al.*, 1973, Eq. 49].

$$K_I = p\sqrt{L\pi} F(L/a) - (\sigma_1 \cos^2 \alpha + \sigma_3 \sin^2 \alpha) F(L/a) \sqrt{L\pi} \\ + (\sigma_1 \cos 2\alpha - \sigma_3 \cos 2\alpha) G(L/a) \sqrt{L\pi} \quad (10)$$

where  $F(L/a)$  and  $G(L/a)$  are given in Table A2 [cf., Paris and Sih, 1965].

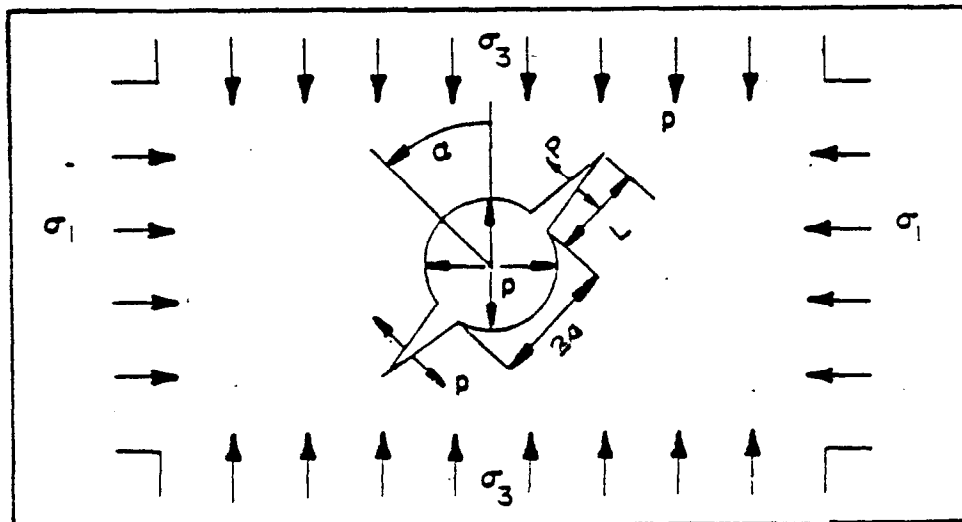


Figure 10. Crack in a borehole wall under far-field stress and internal pressure

In order to determine  $\sigma_1$  from (10) it is necessary to know the minimum compressive stress  $\sigma_3$ , the initial crack length  $L$ , and the breakdown pressure  $p = P_b$  at which the stress-intensity factor  $K_I$  is equal to the critical value  $K_{Ic}$ , required for crack advance. Now  $\sigma_3$  can be determined from measurement of the shut-in pressure,  $P_s$  (see Equation (9)), and  $K_{Ic}$  can be determined from fracture toughness tests (see Appendix). Estimates of  $L$  can be obtained from visual inspection of cores. For given values of  $K_{Ic}$ ,  $P_b$  and  $L$ , Equation (10) becomes a relation between  $\sigma_1$  and  $\alpha$ . This relation can be written in the more useful form

$$-I(\alpha) (\sigma_1 - \sigma_3) = \sigma_3 - P_b + \frac{K_{Ic}}{F(L/a)\sqrt{L\pi}} \quad (11)$$

where

$$I(\alpha) = \cos^2 \alpha - \frac{G(L/a)}{F(L/a)} \cos^2 \alpha \quad (12)$$

The function  $I(\alpha)$  is shown in Figure 11 for various values of the ratio  $L/a$ . Because  $I(\alpha)$  is near zero only for a limited range of values of  $\alpha$ , it appears that (11) can usually be solved uniquely for  $(\sigma_1 - \sigma_3)$  when  $\alpha$  is known.

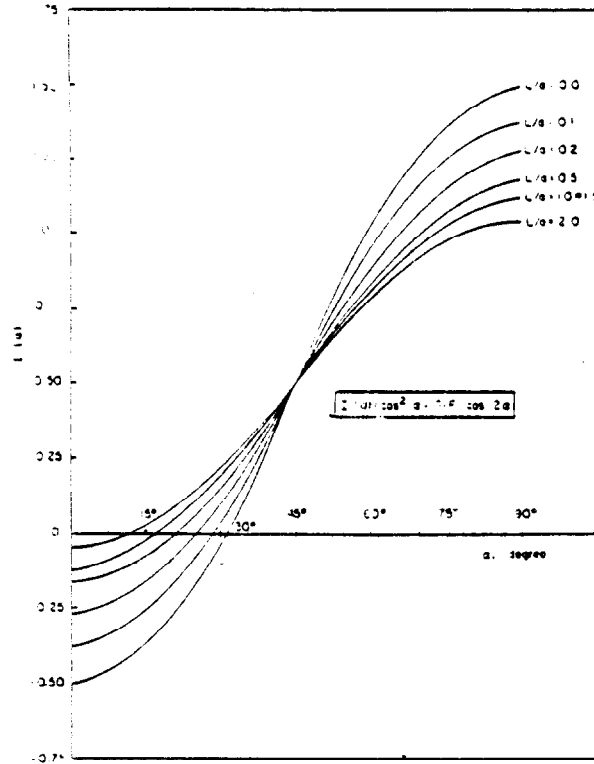


Figure 11.  $I(\alpha)$  as a function of  $\alpha$  for different ratios of  $(L/a)$

If  $\sigma_1 = \sigma_3$ , then the value of  $\alpha$  is irrelevant and the principal stresses in the horizontal plane are

$$\sigma_1 = \sigma_3 = p_b - \frac{K_{Ic}}{F(L/a)\sqrt{L\pi}} \quad (13)$$

On the other hand, the condition  $\sigma_3 < \sigma_1$  requires that  $\sigma_3$  and  $\alpha$  satisfy the inequalities

$$\sigma_3 < P_b - \frac{K_{Ic}}{F(L/a) \sqrt{L\pi}} \quad \text{for } \alpha > \alpha_c \quad (14a)$$

$$\sigma_3 > P_b - \frac{K_{Ic}}{F(L/a) \sqrt{L\pi}} \quad \text{for } \alpha < \alpha_c \quad (14b)$$

where  $\alpha_c$  is the solution of the equation

$$I(\alpha) = 0. \quad (15)$$

Values of  $\alpha_c$  for a wide range of values of  $(L/a)$  (i.e.,  $0 \leq L/a < 2$ ) lie in the interval  $15^\circ < \alpha_c < 30^\circ$ . For known values of  $\sigma_3$ ,  $P_b$ ,  $K_{Ic}$  and  $L$ , inequalities (14) limit the possible range of the values of  $\alpha$ . Unfortunately, however,  $L$  is generally not known with much certainty and allowable variations in the value for  $L$  may lead to satisfaction of either (14a) or (14b). Such uncertainty could be removed if a dominant crack of known length  $L$  could be introduced in the field test.

While it appears that there is no unambiguous way of determining  $\sigma_1$  without additional information, there are a number of observations that can be made. First, if cracks of known length  $L$  could be artificially introduced before hydraulic fracturing of a well bore at two different depths within nominally the same formation, then the principal stress  $\sigma_1$  could be determined without *a priori* determination of  $\alpha$ . This would be accomplished by introducing the cracks at orientations differing by  $\pi/2$  and making use of the identity  $I(\alpha) + I(\alpha + \pi/2) = 1 - G/F$ . Then, from (11)

$$\sigma_1 = \{P_b^\alpha + P_b^{\alpha+\pi/2}\} - \sigma_3 - \frac{2K_{Ic}}{F(L/a) \sqrt{L\pi}} \quad (16)$$

where  $P_b^\alpha$  and  $P_b^{\alpha+\pi/2}$  are the breakdown pressures in the sections with cracks oriented at angles  $\alpha$  and  $\alpha+\pi/2$ , respectively. Once  $\sigma_1$  is obtained



from (16), the angle  $\alpha$  can be obtained from application of (11) to the results of each field test. Comparison of the two values of  $\alpha$  obtained in this way would serve as a check on the validity of the procedure for determining  $\sigma_1$  and  $\alpha$ .

If results from only a single hydrofracturing test are available, then an additional assumption must be introduced in order to estimate  $\sigma_1$ , unless the principal stress directions can be determined from an independent measurement of the hydraulic fracture orientation at large distances from the well bore. In the latter case, measurement of crack orientation at the well bore by means of an impression packer suffices for the determination of  $\alpha$ . If there is no preferred orientation of existing flaws, then the expected value for  $\alpha$  is  $\alpha = 0$  because  $K_I$  is a maximum for  $\alpha = 0$ . On the other hand, even if there is a preferred orientation of existing cracks, the development of these cracks may be such that their orientation will be near  $\alpha = 0$ . Such orientation would result if the paleostresses are still dominant and the cracks are due to extensional failure, i.e., lie in the principal plane of the greatest and intermediate stress [Swolfs, *et al.*, 1976]. In the Devonian Shale core samples the existing vertical cracks are not planar and do not contain pulverized material which would suggest their formation or subsequent sliding in a shear mode. Thus, it appears plausible that initial crack extension in the hydraulic fracturing experiment occurs in a direction near  $\alpha = 0$ . Furthermore, Equation (14b), measured values of  $P_s$ ,  $P_b$ ,  $K_{IC}$  and laboratory values for  $L$  suggest that  $\alpha$  satisfies  $\alpha < \alpha_c$ . Consequently, in what follows, the orientation of the crack at the well bore will be assumed to be  $\alpha = 0$ . This assumption is the same as the

one used in the conventional method for determining  $\sigma_{HMAX}$  discussed previously. Use of the same assumption here facilitates direct comparison of the two approaches.

If one assumes that during a hydrofracturing test the initial crack intersects the borehole in a radial plane perpendicular to the minimum *in situ* compressive stress, then from (11) and (9) one obtains

$$\sigma_1 = \frac{K_{IC}}{\sqrt{\pi L}(G-F)} - \frac{F}{(G-F)} P_b + \frac{G}{(G-F)} P_s \quad (17)$$

where G and F are to be evaluated for appropriate values of L/a.

The critical stress-intensity factor  $K_{IC}$  in (17) is obtained from the results of jacketed pre-notched burst tests as discussed in the Appendix. However, the values of  $K_{IC}$  reported for these tests (Table II) are applicable for fracture from an initial state of zero confining pressure, whereas, the pre-fracture state in the field involves substantial confining pressures. Recent experiments on the fracture toughness of Indiana Limestone [Schmidt and Huddle, 1976 and Abou-Sayed, 1977] indicate that the critical stress-intensity factor at which crack extends increases significantly with the increasing confining pressure. Although the dependence of  $K_{IC}$  on confining pressure has not been measured for Devonian Shale, it appears likely that the value of  $K_{IC}$  for shale will be considerably larger under the *in situ* confining pressure than measured in the unconfined laboratory tests. In the absence of data on the effects of confining pressure on  $K_{IC}$  for shale from the test section, it appears reasonable to assume that  $K_{IC}$  increases by a factor of 1.6--the fractional increase of  $K_{IC}$  reported for Indiana Limestone under a confining pressure of 24 MPa (3500 psi) corresponding to the mean *in situ* normal stress at the test section.

The crack length  $L$  in (17) is taken to be 8mm. This length corresponds to the length of most pre-existing vertical cracks observed in the test cores (see Figure 6). This choice appears reasonable because many such cracks can be expected to intersect the 22.2 cm hole and it is likely that for at least one of these cracks the depth is essentially equal to the initial length of the crack.

The dimensionless stress-intensity factors  $F$  and  $G$  should be reduced because the cracks are not plane-strain cracks. The extent of the reduction is difficult to assess, but comparison of plane-strain-crack solutions with solutions for a circular crack in an infinite medium, [Sneddon, 1946], and elliptical crack in a plate, [Rice and Levy, 1970] or hollow cylinder [Underwood, 1972] suggests that the stress-intensity factor could be reduced by approximately 40 percent from the value for a plane-strain crack of the same depth. The functions  $F$  and  $G$  should decrease similarly so that quotients  $F/(G-F)$  and  $G/(G-F)$  should be essentially the same as for plane-strain conditions. The quantity  $(G-F)$  in the first term of (17) can be expected to be approximately 60 percent of the value for the plane-strain conditions.

Incorporating these considerations in the application of (17) one finds that the maximum horizontal compressive stress  $\sigma_{HMAX} = \sigma_1$  is approximately

$$\sigma_1 \approx \frac{38.2 \times 1.6}{\sqrt{\pi} 8 (2.93 - 2.06) .60} - \frac{2.06}{.97} (20.2) + \frac{2.93}{.87} (16.3)$$

or

$$\sigma_1 \approx + 23.2 - 47.8 + 54.9$$

or

$$\sigma_{HMAX} = \sigma_1 \approx 30.3 \text{ MPa (4390 psi)}. \quad (18)$$

The intermediate steps are shown in order to indicate the relative magnitude of the terms. The accuracy in the computed value of  $\sigma_{HMAX}$  is reduced because the computation involves differences in terms of comparable magnitude. However, the two large terms are regarded as known with quite good certainty (say  $\pm 5$  percent) and the larger percent error (say  $\pm 50$  percent) occurs in the smallest term. As a result, the probable error in the computed value of  $\sigma_{HMAX}$  is regarded as about 40 percent. The calculated values of the principal *in situ* stresses based on a linear elastic fracture mechanics analysis are summarized in Table IV along with other relevant parameters.

TABLE IV

The *In Situ* Stress at 837m (2745 feet) Based  
on Linear Elastic Fracture Mechanics

Depth (h) m (feet)	$P_b$ MPa (psi)	$P_s$ MPa (psi)	L mm (in.)	$\sigma_{OB}$ (Vertical) MPa (psi)	$\sigma_{HMAX}$ MPa (psi)	$\sigma_{HMIN}$ MPa (psi)	Direction of $\sigma_{HMAX}$
837 (2745)	20.2 (2930)	16.3 (2360)	8	22.1 (3210)	30.3 (4390)	16.3 (2360)	N45°-50°E

In order for the computations shown in (18) to be consistent with a model of breakdown based on (10) it is necessary for the entire length of the dominant well bore crack to be subjected to the breakdown pressure  $P_b$ . For tight natural cracks of the type observed in the cores, pressurization of the crack faces will probably not occur unless the crack is opened by circumferential tensile stress at the wall of the bore. From (1), this circumferential tensile stress is given by

$$\sigma_{\theta\theta}|_{r=a} = \sigma_{HMAX} - 3P_s + P_b + P_o$$

which for the values given in Table IV, yields  $\sigma_{\theta\theta}|_{r=a} = 3.3$  MPa (480 psi).

Because the crack faces cannot support a tensile stress, the faces should separate and allow the low viscosity pressurizing fluid to penetrate. Once

the fluid begins to penetrate, the crack should open further and allow the fluid to advance until the fluid pressure acts over essentially the entire surface area of the crack. In view of the combination of slow pressure build-up and low fluid viscosity, the effects of fluid flow on breakdown [Zoback and Pollard, 1977] are not expected to be significant in this case.

The primary uncertainties in the calculation of  $\sigma_{HMAX}$  are associated with the geometry of the dominant well bore crack and with the value of  $K_{IC}$  for the loading conditions of the field test. The uncertainty regarding the crack geometry can largely be removed by cutting vertical cracks of known depth  $L$  as discussed in connection with Equation (16). Uncertainty associated with the value of  $K_{IC}$  can be reduced by employing laboratory fracture toughness experiments which simulate the stress state existing in the field. Such experiments can be conducted employing the configuration shown in the Appendix with independent control of axial stress, confining pressure, and internal pressure. Also, the influence of errors in  $K_{IC}$  on the computed value of  $\sigma_{HMAX}$  can be reduced by an optional choice of  $L$  such that the first term in (17) and (18), i.e., the term that depends on  $K_{IC}$  is made as small as possible.

The latter concept of reducing the relative significance of terms related to rock strength is similar to the concept [Bredehoeft, *et al.*, 1976] of employing cyclic pressurization of the well bore to determine the maximum compressive stress  $\sigma_{HMAX}$ . In this approach the first cycle of pressurization to breakdown is assumed to propagate a vertical crack in the direction perpendicular to the minimum compressive stress  $\sigma_3$  so that breakdown in subsequent cycles of pressurization is associated with re-opening the induced crack. Then the *in situ* stress  $\sigma_{HMAX}$  is computed from (1) with  $T = 0$  and  $P_b$  equal to the breakdown pressure for the second and later cycles

of pressurization. The cyclic pressurization technique has the obvious advantages of simplicity and accuracy through elimination of rock strength parameters that are the least well-known terms in (1) and (17). This technique holds considerable promise, provided that breakdown on subsequent pressurization cycles can be related unambiguously to the opening of induced cracks perpendicular to the minimum compressive stress.

#### Comparison of Two Methods for Computing $\sigma_{HMAX}$

So far, two methods for computing  $\sigma_{HMAX}$  have been discussed. The first, based on a critical tensile stress fracture criterion and neglect of consideration of cracks in the wall of the laboratory specimen and the field well, gives [from (1) and (4)]

$$\sigma_{HMAX}^t = 3P_s - P_b + \left(\frac{w^2+1}{w^2-1}\right) P_i - P_o. \quad (19)$$

The second, based on linear elastic fracture mechanics gives

$$\sigma_{HMAX}^f = \frac{G}{(G-F)} P_s - \frac{F}{(G-F)} P_b + \frac{K_{Ic}}{0.6(G-F) \sqrt{\pi L}} \quad (20)$$

For both (19) and (20) failure is assumed to occur on the radial plane perpendicular to the minimum *in situ* compressive stress.

If  $L/a$  is much less than unity, say less than 0.1, then  $G$  is approximately 1.5  $F$  and  $G-F$  is approximately unity so that (20) reduces to

$$\sigma_{HMAX}^f = 3P_s - 2P_b + \frac{K_{Ic}}{\sqrt{\pi L} (0.6)} \quad (21)$$

Comparison of (19) and (21) reveals the primary differences between the two methods for computing  $\sigma_{HMAX}$ . The term in (21) related to the breakdown

pressure  $P_b$  is twice the corresponding term in (19). This doubling of the effect of the bore pressure  $P_b$  is due to the fact that, according to linear elastic fracture mechanics, the stress-intensity factor for a shallow, pressurized, radial crack in the wall of a cylindrical cavity is obtained from the addition of two equal effects: (i) the circumferential tensile stress  $P_b$  due to internal pressure acting on the well bore and (ii) the pressure  $P_b$  acting on the crack face. Superposition, locally, of a uniform tensile stress  $P_b$  removes the compressive stress on the crack face and causes the region containing the shallow crack to be subjected to an effective tensile stress  $P_b$ , in addition to the tensile stress  $P_b$  due to (i). The effect (ii) of pressure acting on the faces of the well bore crack is not included in (19).

The second difference between (19) and (21) is in the treatment of rock strength. In (21) the rock strength is characterized by the size  $L$  of the dominant crack and by the critical stress-intensity factor  $K_{IC}$  necessary for crack advance. In (19) the rock strength is characterized by the nominal tensile stress in the wall of a thick-walled cylinder that bursts under an internal pressure  $P_i$ . Because  $P_i$  and  $P_b$  are generally not equal, the influence of pressure acting on bore hole cracks is not incorporated consistently when (19) is used. Also, Equation (19) does not incorporate the size effect associated with the probability that the dominant crack intersecting the well bore will be larger than the largest crack intersecting the bore of the laboratory specimen.

Finally, use of (19) requires the value of the pore pressure  $P_o$  at breakdown. On the other hand, the pore pressure does not appear in (21) but is incorporated directly because, according to linear elastic fracture mechanics, the stress-intensity factor for a given geometry and loading is

unaffected by superposition of a uniform hydrostatic pressure. Thus, if the total stress  $\sigma_{HMAX}^f$ ,  $P_s$ , and  $P_b$  in (21) are replaced by  $\tilde{\sigma}_{HMAX}^f + P_o$ ,  $\tilde{P}_s + P_o$ ,  $\tilde{P}_b + P_o$ , where  $\tilde{\sigma}_{HMAX}^f$ ,  $\tilde{P}_s$ , and  $\tilde{P}_b$  denote effective stresses, then (21) is changed only by the replacement of total stresses by effective stresses and pore pressure does not appear explicitly. The pore pressure is required in (19) in order for a similar substitution to lead to a relationship between effective stresses and rock strength that is independent of pore pressure.



## CONCLUDING REMARKS

A fracture mechanics analysis of hydrofracturing indicates that, if the principal stresses are not equal, fractures will tend to become oriented normal to the minimum *in situ* compressive stress as the crack length increases. For crack lengths of several meters the crack should be quite nearly perpendicular to the minimum compressive stress and the shut-in pressure  $P_s$  should provide an accurate estimate of the minimum principal stress. These conclusions are consistent with previous interpretations of hydrofracturing tests.

The fracture mechanics approach provides helpful insight into the more difficult problem of determining the principal stress directions and the maximum principal stress. This approach suggests that more accurate determination of the maximum principal stress can be achieved if better information is obtained on the size and shape of the dominant well bore crack and on the value of  $K_{IC}$  under field conditions. The analysis suggests that uncertainties associated with crack geometry can be reduced greatly by modifying hydrofracturing tests to include the cutting of narrow notches of known depth. If these notches are made deeper than all cracks in the cores, then the hydrofracturing test should become comparable to that of the laboratory test used for determining  $K_{IC}$ . Also, by notching two different sections in two known orthogonal directions it should be possible to determine  $\sigma_{HMAX}$  and the principal stress directions with reasonable certainty (See Equation (16) and the discussion following it). Improved values of  $K_{IC}$  under field conditions should be attainable by conducting fracture toughness tests under confining pressure as discussed by Abou-Sayed [1977].

The measured principal stress directions agree well with the prevailing geological structure in the region and with the principal stress directions reported by Overbey [1976] from a series of measurements in West Virginia. A basement structure map of the region in which the reported test was conducted is shown in Figure 12. The map, adapted from Harris (USGS Map I-919 D, 1975) by Overbey [1976], shows a projection of the Rome Trough through the northwestern edge of West Virginia. Schumaker [1976] describes the Rome Trough as a graben bounded by high-angle, normal faults. This structure lies within an area which is a junction of three distinct geological provinces [Werner, 1976].

- 1) Central Appalachian Fold Belt
- 2) Southern Appalachian Thrust Fault Belt
- 3) Appalachian Plateau with Basement Faults

Although it is not known whether or not the basement faults penetrate into the Devonian Shales in this area, Overbey's measurements [1976] suggest a correlation of principal stress directions with the basement structure. At locations which fall outside the projection of the Rome Trough, the measured direction of  $\sigma_{HMAX}$  are oriented generally E-W and are normal to the predominately N-S strike of the thrust faults and foldings of the Appalachian Mountains. Measured directions of  $\sigma_{HMAX}$  which fall within the projection of the Rome Trough trend N 45° E to N 50° E, or parallel to the strike of the basement faults and in agreement with the directions reported here.

For the relative magnitudes of the principal stresses to be consistent with the expected state in a region bounded by normal faults [Hubbert and Willis, 1957] the vertical stress should be the maximum principal stress. The minimum principal stress direction should be normal to the strike of the

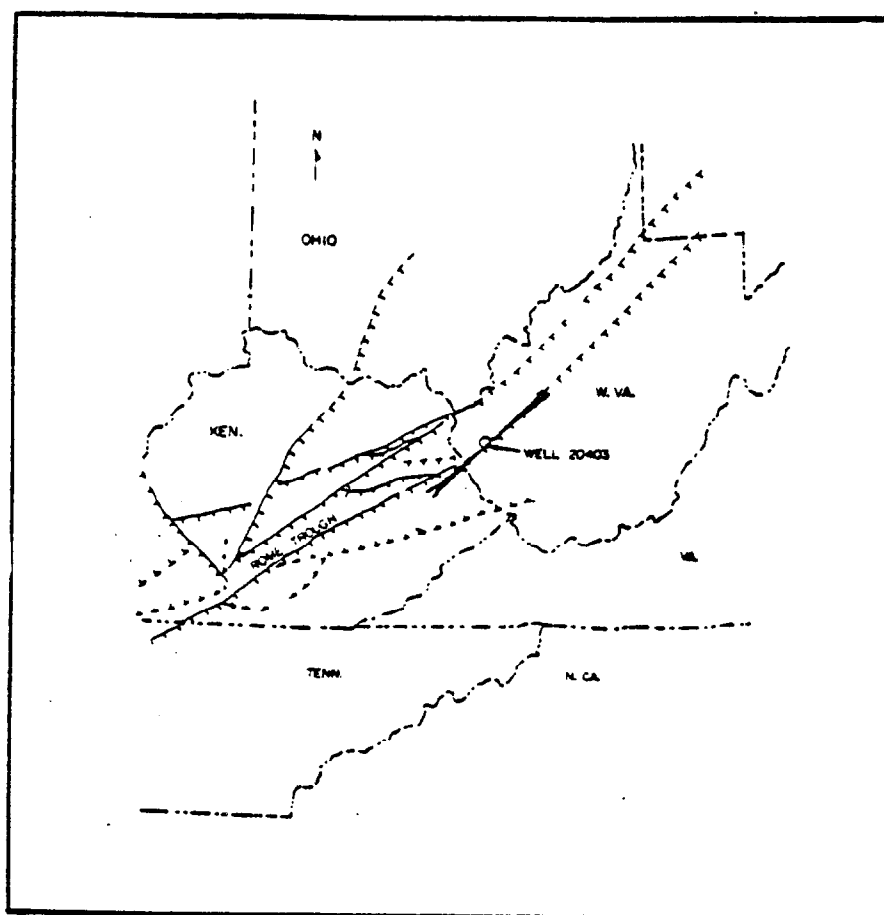


Figure 12. Basement Structure of Kentucky-West Virginia Adapted from Overbey [1976] with the Direction of Maximum Horizontal Stress at Well 20403

normal faults and the intermediate principal stress direction should be parallel to the strike of the faults. In order for the *in situ* stresses reported here to conform with such an orientation of principal stress directions it would be necessary to reduce the estimated value of  $\sigma_{HMAX}$  to a value between  $\sigma_{HMIN}$  and  $\sigma_{OB}$ . According to linear elastic fracture mechanics such a revision in the computed value of  $\sigma_{HMAX}$  would require (i) an initially open well bore crack that could accept the fracturing fluid at a pressure  $P_D$  for which the nominal circumferential stress at the well bore is compressive and (ii) an appreciably larger crack length

and/or smaller  $K_{IC}$  so that the first term in (18) can be reduced by roughly a factor of two. Although this combination of circumstances is clearly possible, it is believed to be more likely that the maximum principal stress is horizontal as reported in Table III and Table IV.

In summary, the complex geological structure in this region does appear to be related to the measured orientation of the stress field. It is not yet clear why stresses relating to the large scale deformation and mountain building during the Paleozoic Era should still remain a controlling factor in this region. However, the evident influence of ancient structural features on present-day measurements should discredit attempts to smooth these measurements across distinct structural boundaries.

## APPENDIX

## Description of Laboratory Tests

Fracture-Toughness Tests

The experiment used to determine a materials resistance to fracture extension consists of subjecting a prenotched, thick-walled cylinder to an internal pressure that loads the wall at the inner radius but does not load the faces of the notch. This type of test has been proposed by Clifton, et al., [1976] to measure the critical stress-intensity factor  $K_{IC}$ , of geologic materials. The experimental configuration is shown in Figure A1. The test specimen is a thick-walled cylinder with an outer-to-inner-diameter ratio of 10 or more and an outer diameter of 96.5mm. Two radially opposed prenotches are cut into the inner bore and penetrate one tenth of the wall thickness along the full length of the specimen. Prenotching is accomplished using a diamond-impregnated copper wire, 0.2 mm (0.008 in.) in diameter. A soft impermeable jacket of urethane tubing with a thickness of 0.15mm (0.060 in.) prevents the fluid pressure from loading the crack faces and the fluid from permeating the rock matrix. The jacket extends into steel end caps that are attached to each end of the sample. The seal for the pressurizing fluid is made in the steel end caps as shown in Figure A1. A cone shaped rubber plug forms a seal between the inside of the urethane jacket and the fluid port when a small axial force is applied to the system. The seal pressure is varied by changing the amount of interference between the steel end cap and the compression rod. A small steel tube in the upper rubber end seal allows fluid to communicate between the inner bore and the high pressure line.

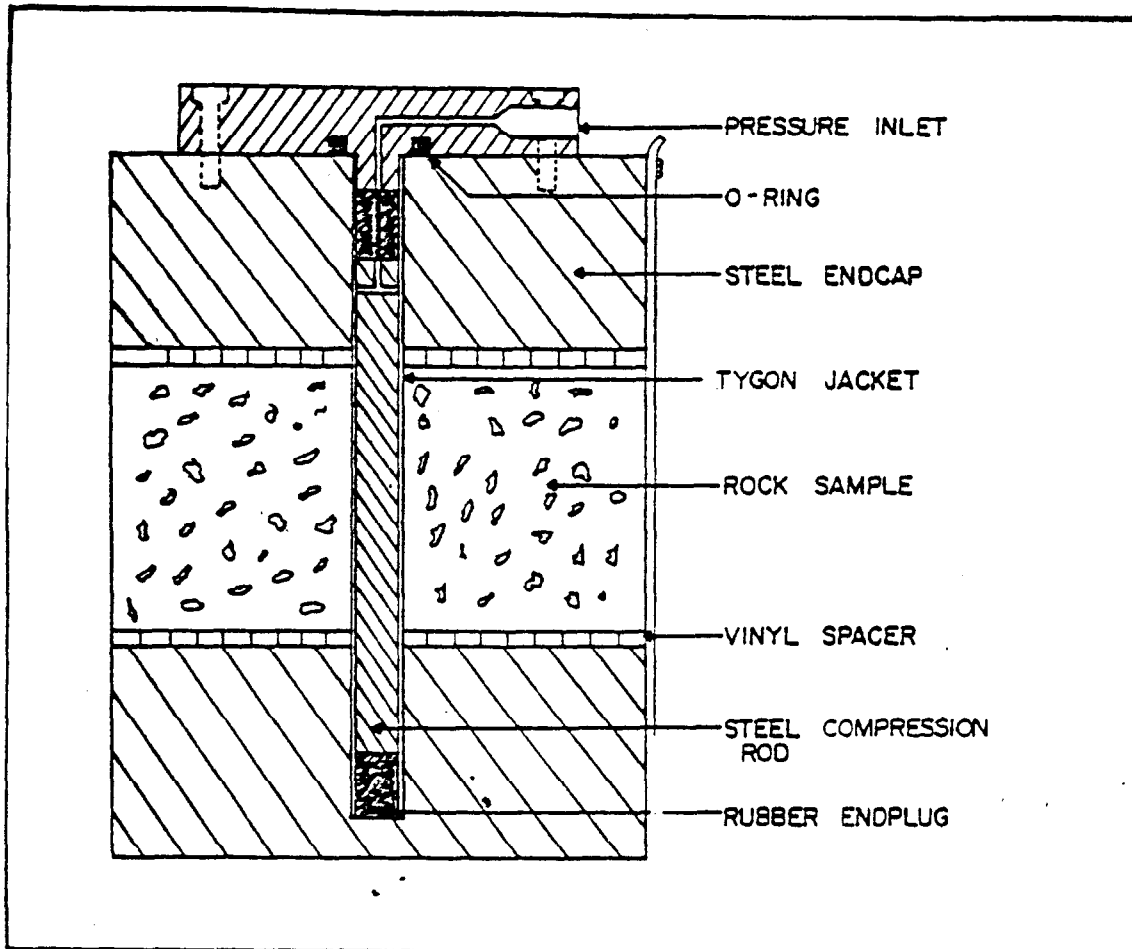


Figure A1. Cross-section of a typical specimen for fracture-toughness-burst test.

To evaluate  $K_{IC}$  for a given sample material the internal pressure is increased slowly and recorded until a crack is observed to propagate. As predicted by the analysis of this configuration [Bowie and Freese, 1970] the first phase of crack propagation is stable for wall-thickness ratio greater than 10. Further increase of the internal pressure initiates unstable crack growth and results in catastrophic failure of the cylindrical specimen. The maximum internal pressure ( $P_m$ ) that can be applied to the inner wall is related to the critical stress-intensity factor,  $K_{IC}$ , of

the tested material and the geometry of the sample, by the following equation:

$$K_{IC} = K_{IC}^* (P_m \sqrt{\pi a_0})$$

where  $a_0$  is the internal radius and  $K_{IC}^*$  corresponds to the local minimum value of the numerically calculated function  $K_I^*$  shown in Figures A2 and A3 for different wall thickness ratios,  $w$ , and crack geometry. For example, for a wall thickness ratio of 10,  $K_{IC}^*$  is given by 0.245 or 0.42 depending on whether the sample has single or double radial cracks, respectively.

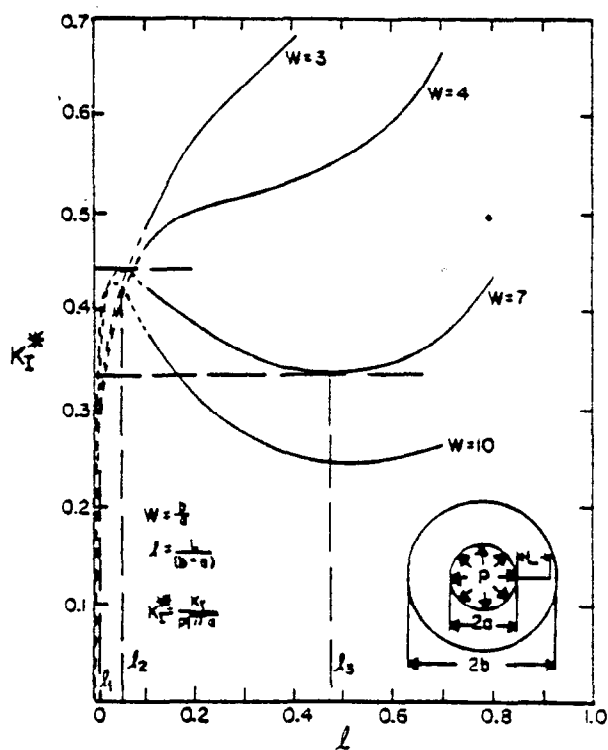


Figure A2. Stress-intensity factor for jacketed cylinder with one radial crack (from Bowie and Freese, 1972).

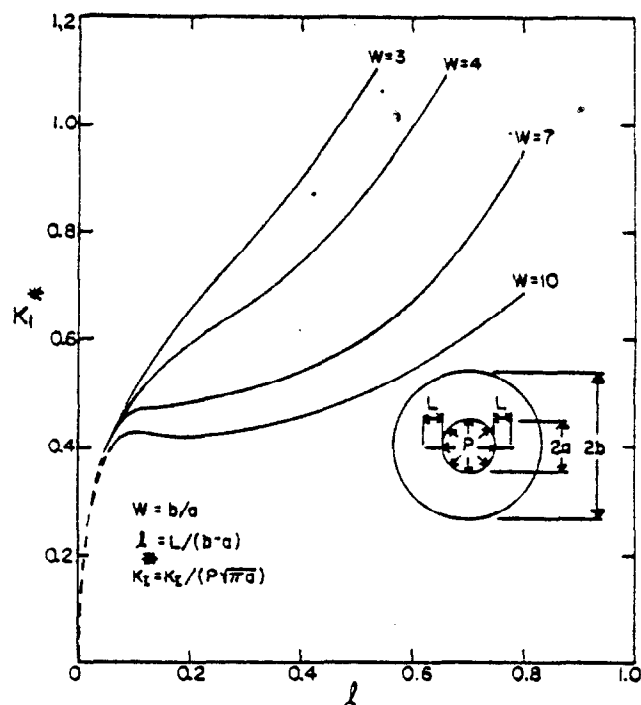


Figure A3. Stress-intensity factor for jacketed cylinder with two radial cracks (from Bowie and Freese, 1972).

### Results of Fracture Toughness Tests

The results of the fracture toughness tests are further evidence of the anisotropic nature of this shale (see Table A1). Figures A4 and A5 are photographs of the two samples that were pre-notched and pressurized with impermeable membranes. In Figure A4 the failure plane originated at the tips of the pre-notch but turned the moment it intersected a natural fracture. The failure plane in Figure A5 began in three places, at one pre-notch and at radially opposed points along a natural fracture, however, catastrophic failure occurred along the natural fracture.

TABLE A1  
Results of Fracture Toughness Tests

Sample Depth m (ft.)	$b_o/a_o$	$P_m$ Failure Pressure MPa (psi)	$K_{Ic}$	$K_{Ic}$ MPa/√m (psi/√in)	Remarks
826.5 (2711)	10.47	19.5 (2825)	0.4	42.4 (1220)	Failure initiated along notch then turned along pre-existing fracture
839 (2761)	10.47	26.21 (3800)	0.25	38.2 (1100)	Failure ignored notch and occurred on a pre-existing fracture

TABLE A2  
Tabulation of Functions F and G [Paris and Sih, 1965]

L/a	One Radial Crack		Two Radial Cracks	
	F(L/a)	G(L/a)	F(L/a)	G(L/a)
0.00	2.26	3.39	2.26	3.39
0.10	1.98	2.73	2.06	2.93
0.20	1.82	2.30	1.83	2.41
0.30	1.69	2.04	1.70	2.15
0.40	1.58	1.86	1.61	1.96
0.50	1.49	1.73	1.57	1.83
0.60	1.42	1.64	1.52	1.71
0.80	1.32	1.47	1.43	1.58
1.00	1.22	1.37	1.38	1.45
1.50	1.06	1.18	1.26	1.29
2.00	1.01	1.06	1.20	1.21
3.00	0.93	0.94	1.13	1.14
5.00	0.81	0.81	1.06	1.07
10.00	0.75	0.75	1.03	1.03
-	0.707	0.707	1.00	1.00



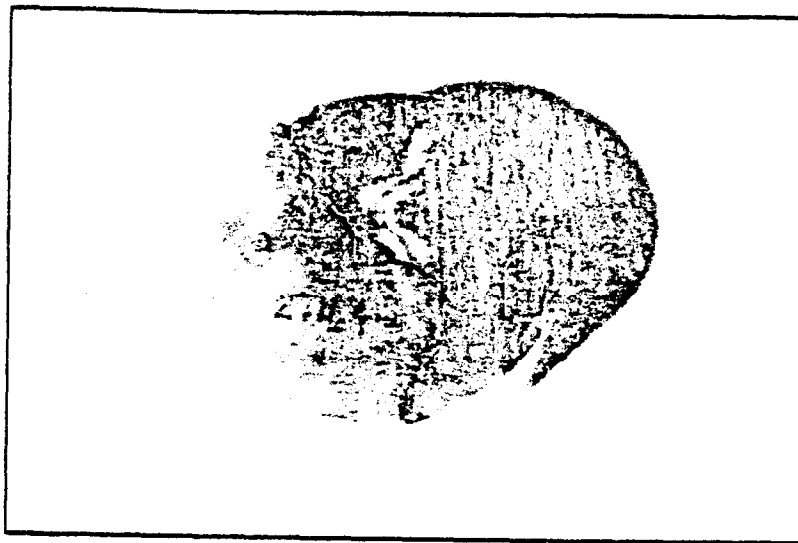


Figure A4. Fracture toughness sample from 826.5 m (2711 ft.)

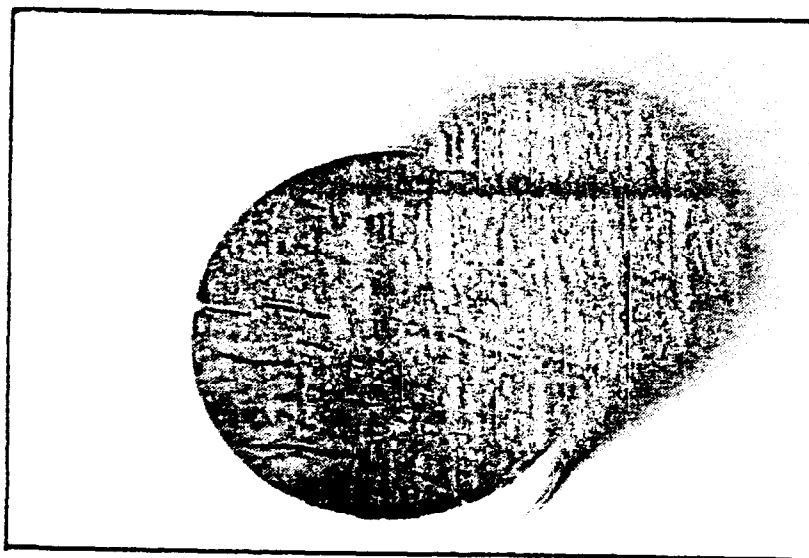


Figure A5. Fracture toughness sample from 829 m (2761 ft.)

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TERRA TEK REPORT

DETERMINATION OF THE STRAIN RELAXATION AND THEIR  
RELATION TO SUBSURFACE STRESSES IN THE DEVONIAN SHALE

FINAL REPORT  
TASK II

DETERMINATION OF THE STRAIN  
RELAXATION AND THEIR RELATION TO SUBSURFACE  
STRESSES IN THE DEVONIAN SHALE

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## SUMMARY OF TASK II

DETERMINATION OF THE STRAIN  
RELAXATION AND THEIR RELATION TO SUBSURFACE  
STRESSES IN THE DEVONIAN SHALE

Strain-relaxation tests were performed on twenty-four specimens of Devonian Shale obtained from the Columbia Well No. 20402, Lincoln County, West Virginia, for the overall purpose of identifying, among the gas-bearing shales, the prime zones for stimulation (MHF) treatment within the Devonian Shale sequence.

The results of this work provide information on the strains and stresses in only three shale zones - Upper Gray Shale, Middle Gray Shale and Middle Brown Shale - and yield the following interpretations:

1. The rate of strain relaxation is 2 to 2.5 times higher in the Middle Brown Shale zone than in the overlying Gray Shale zones, which suggest that
2. The minimum horizontal matrix-stress in the Middle Brown Shale zone could be up to 30 percent lower than the minimum horizontal matrix-stress in the upper Gray Shale zones.

From these measurements alone it appears that the upper Gray Shale zones may act as an effective barrier which prevents upward fracture propagation, thereby promoting larger lateral extension of a fracture initiated within the Middle Brown Shale, a prime gas-bearing zone.

## INTRODUCTION

The question addressed in this report is as follows: Can an inexpensive but reliable core-analysis technique be developed to provide useful information on the present-day stress distribution in subsurface reservoirs and aid in the identification and selection of prime gas-bearing zones for stimulation by massive-hydraulic-fracture (MHF)?

To provide answers to this question we have begun to use a technique that very accurately measures the small dimensional changes (strain relief) of Devonian Shale samples after they are cored and brought to the surface. The smallest dimensional change (strain) that can be detected by this technique is of the order of several micro-inches\*. For example, a four-inch diameter shale sample that strained or relaxed ten micro-inches ( $\Delta l/l = 10 \times 10^{-6}$ ) changed dimensions by 0.00004 inch.

Basically, the idea is that as a sample of any kind of rock is cored and taken from its subsurface environment, it will experience a rapid change in stresses and consequently relax and change dimensions. The amount of relaxation over a period of time after coring is related to the magnitude of the subsurface stresses or, more specifically, the magnitude of the difference between the greatest (overburden) stress and the least (minimum-horizontal) stress in the formation from which the sample was obtained.

From these relaxation measurements, however, we can only develop a qualitative estimate of the distribution of the subsurface minimum-horizontal stress-gradient and determine the relative changes, if any, in the gradient distribution from formation to formation. Nevertheless, this kind of information is useful and important because it can lead to an early recognition of anomalies in the subsurface stress gradients (usually assumed

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\* One micro-inch = one inch per one million inches =  $10^{-6}$ .

to increase linearly with depth) and the identification of prime zones for stimulation and fracture-treatment.

In this report we will briefly outline the important aspects of the well-site technique of strain relaxation and its limitations. These will be followed by the description of the strain-relaxation results obtained in samples from the upper Gray Shale and Middle Brown Shale zones in Columbia Well No. 20402. Next, attention will be focused on how these results can be interpreted to provide new and useful information and, finally, how they can be applied and lead to improvements in the results of a fracture treatment program.

## WELL-SITE TECHNIQUES

Strain Relaxation

The technique used at the well-site consisted of selecting small pieces of rock as soon as they were removed from the core barrel and laid out for initial geologic identification and description. The rock was slabbed with a rock saw to provide two flat, parallel surfaces and result in a test specimen two inches thick and four inches in diameter. The top surface was air dried and then washed with acetone to dissolve whatever moisture was retained at the surface of the sample. A 45-degree strain-rosette was fastened in the central part of the top surface using a quick-setting epoxy (Figure 1). Gage No. 2 (center gage) in each rosette was aligned with the reference (orientation) groove of the sample. The principle of operation of a 45-degree strain rosette is outlined in Appendix I.



Figure 1. Core sample instrumented with 45-degree strain rosette. Gages 1, 2 and 3 are seen from left to right. The sample is Dakota sandstone cored in a western gas well.

Each of the three strain gages in the 45-degree rosette were wired into a switching unit and a strain indicator. The switching unit allowed the concurrent measurement of strain changes in as many as 12 specimens. Data were recorded manually every half-hour for the first day and every hour for all subsequent days. Three strain-time curves were drawn for each specimen to provide the basic data from which to calculate the rates and magnitudes of strain relaxation.

The limitations in reliably measuring the strain relaxation in rock samples should be clearly recognized. They are listed as follows:

1. The rock sample must be representative of the subsurface formation.
2. The rock samples must be maintained under constant moisture and temperature conditions. A change in temperature of  $1^{\circ}\text{F}$ , for example, will result in an apparent strain of about  $6 \times 10^{-6}$  or 6 micro-inches.
3. The rock samples should be oriented geographically to determine the directions of the major axes of strain relaxation.
4. The rock sample should be instrumented soon after removal of the subsurface stresses to assure maximum accuracy of measurement.
5. The elastic strain-relaxation, which occurs instantaneously upon coring down-hole, is not detected by this technique.

#### Velocity Measurements

Longitudinal (P) wave measurements were made on rock samples as soon as possible after the coring of Columbia Well No. 20402. These measurements were made in conjunction with, and with the same objective as, the strain relaxation measurements.

Figure 2 shows the technique used to obtain the transit times of the ultrasonic wave through the rock sample. A DC pulse generator was used to

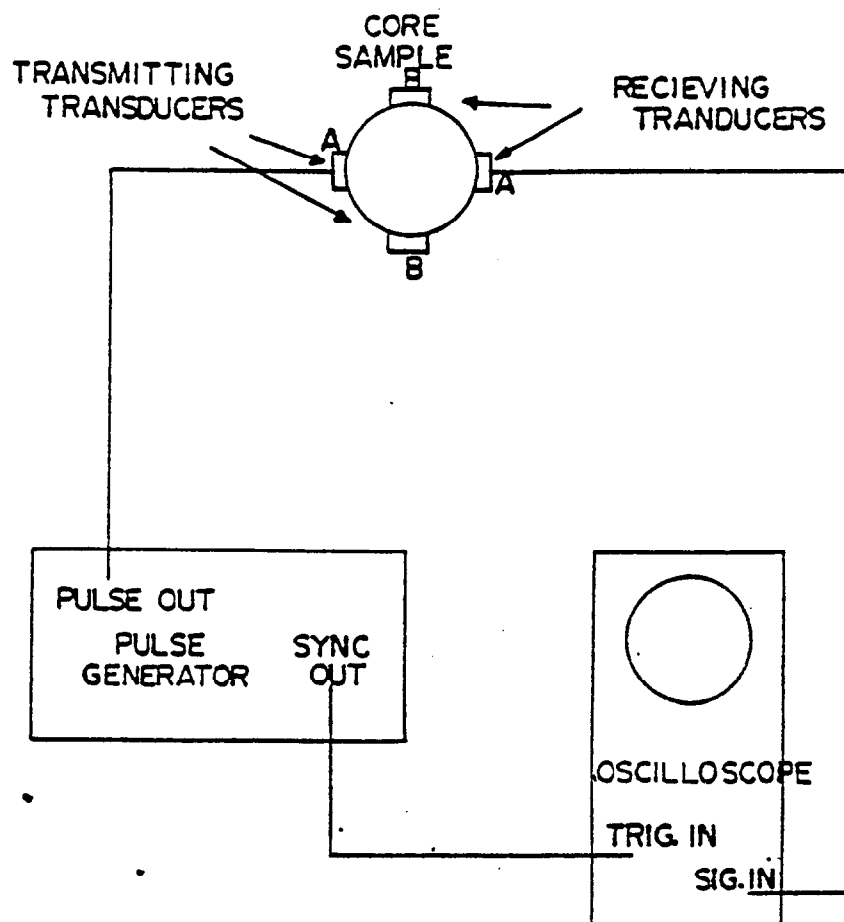


Figure 2. Schematic diagram of field equipment used to measure changes in P-wave velocity along two horizontal directions in a relaxing core.

excite the transmitting transducer and the same time to initiate the sweep on an oscilloscope. The signal produced by the receiving transducer was displayed as a vertical deflection on the oscilloscope trace. The travel times were obtained directly from the calibrated sweep speed of the trace. The accuracy of the time measurements was of the order of three percent.

The transit times were determined in two planes across the rock sample. In order to eliminate the effects of changes in the transducer to sample bonding the transducer assembly (Figure 3) was firmly attached to the rock,

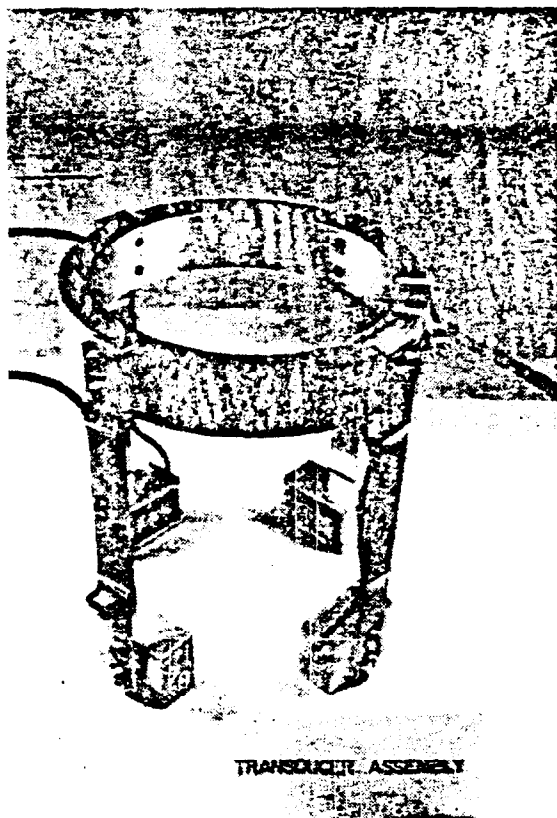


Figure 3. Photograph of sonic-transducer assembly.

and remained so for the duration of the test. The transducer elements were made from PZT-5 material and were cut to resonate at a natural frequency of 200 kilohertz.

It should be emphasized, however, that significant changes in velocity across the rock sample will occur only if the strain relaxation in the rock sample exceeds a certain minimum value, thereby causing significant changes in the elastic moduli of the rock sample. In some sandstones, for example, the lower limit of strain relaxation is about 500 micro-inches ( $5 \times 10^{-4}$ ) beyond which velocity changes can be observed.



## RESULTS

### Strain Relaxation

The strain-time plots (Figures A1 through A21, Appendix II, form the basic data and consist of three curves. Each curve represents each strain gage in the 45-degree rosette applied to the rock surface (Figure 1). Because of the large, daily temperature variations at the well site during the measurement period, each curve has been corrected for temperature using a correction factor of  $10 \times 10^{-6}/^{\circ}\text{C}$  ( $6 \times 10^{-6}/^{\circ}\text{F}$ ).

The data indicates that the very small strain changes with time are nearly uniform in all directions. There were a few exceptions, however. Samples No. 1 (Figure A1) showed maximum horizontal, relaxation of about  $240 \times 10^{-6}$  along a direction (N  $10^{\circ}$  W) almost perpendicular to the predominant fracture trend (N  $70^{\circ}$  E) in the Devonian Shale. The most dramatic strain changes were observed in Sample No. 7 (Figure A4). The overall relaxation, nearly uniform in the horizontal plane ( $\epsilon_{\text{max}} = 400 \times 10^{-6}$ ;  $\epsilon_{\text{min}} = 350 \times 10^{-6}$ ), is due to the formation of a sub-horizontal fracture about 1.5 inches below the top of the specimen during the time of measurement. Velocity measurements (Table I, 2768 feet), made below the fracture, indicated no measureable changes in P-wave velocity with time.

Nearly uniform or isotropic strain-relaxation was typical in the remainder of the shale specimens. Table I summarizes this information and also lists the relaxation rates (micro-inches per hour) for each specimen. Because at least three specimens were sampled from the bottom portion of the well, the relaxation rates in each group of samples were averaged and plotted against depth in Figure 4. This plot clearly shows that the relaxation rates are 2 to 2.5 times higher in the Middle Brown Shale (about  $4$  to  $5 \times 10^{-6}$ /hour) than those measured in the Upper and Middle Gray Shales (about  $2 \times 10^{-6}$ /hour).

### Velocity Measurements

Ultrasonic measurements were made on a total of twelve core segments. The transducer assembly was mounted to the samples as soon as possible after coring, without interfering with the strain gaging operation. Table II is a listing of the changes in transit times at various intervals after the initial readings. The changes in transit time in all of the samples are very slight and are not considered in the discussion that follows.

TABLE I  
Strain Relaxation Tests

Sample (#)	Depth (feet)	Strain Relaxation Gage 1 2 3 ( $\times 10^{-6}$ )			Elapsed Time (hours)	Average Relaxation per Sample ( $\times 10^{-6}$ )	Average Relaxation Rate per Sample ( $\times 10^{-6}$ /hour)	Average Relaxation Rate per Core Run ( $\times 10^{-6}$ /hour)*
3	2709	40	50	60	14	50	3.6	1.8
6	2703	0	0	0	14	0	0	
8	2759	30	0	0	15	10	0.7	2.1
9	2746	110	20	30	15	53	3.5	
10	3058	50	10	25	40	28	0.7	2.0
11	3057	150	200	150	40	167	4.2	
12	3049	50	50	35	40	45	1.1	
13	3117	25	15	10	24	17	0.7	1.2
14	3116	0	25	30	24	18	0.8	
15	3114	75	50	35	24	53	2.2	
16	3348	175	165	130	30	157	5.2	4.0
17	3343	40	110	120	30	90	3.0	
18	3340	85	130	135	30	117	3.9	
19	3403	150	110	60	32	107	3.3	4.0
21	3399	140	170	150	32	153	4.8	
22	3458	80	130	90	16	100	6.3	5.7
23	3457	90	110	100	16	100	6.3	
24	3435	70	60	80	16	70	4.4	

\* This column is plotted against depth in Figure 25.

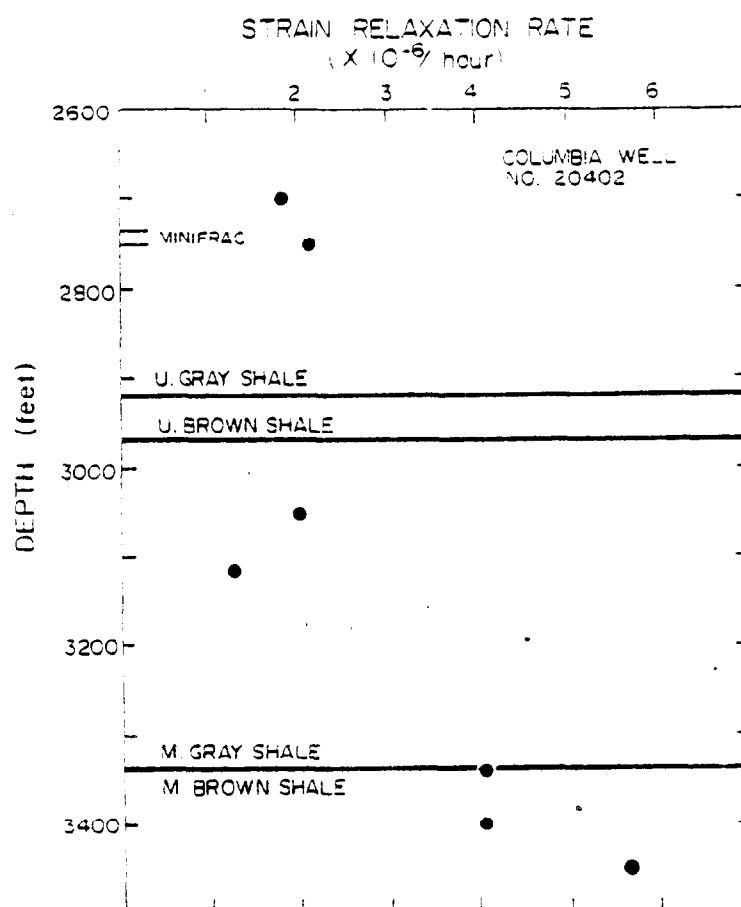


Figure 4. Plot of strain-relaxation rate against depth in Columbia Well No. 20402.

TABLE II  
Change in P-Wave Transit Times with Elapsed  
Time on Cores Retrieved from Columbia Well #20402

Sample Depth (ft)	Elapsed Time		Normalized Transit Time (t/to)	
	Hrs.	Min.	"A"	"B"
2586		30	0.97	1.00
2586	14	10	0.97	0.98
2586	16	25	1.00	0.98
2698		30	1.00	1.00
2709		15	1.00	1.00
2709	11	30	1.02	1.00
2768		20	1.00	1.00
2768		50	1.00	1.00
2768	3	50	1.01	1.00
2768	26	20	1.01	1.00
3049		15	1.00	1.00
3049		35	1.00	1.00
3049	1		1.00	1.00
3049	13	50	0.98	1.00
3114		20	1.01	0.98
3114		40	1.03	0.98
3114	1	30	1.03	0.98
3114	24	10	1.01	0.98
3117		20	1.00	0.99
3117		50	1.00	0.97
3117	1	35	1.00	0.97
3117	24	20	0.97	0.97
3340		10	1.00	1.00
3340		40	1.01	1.01
3340	1		1.03	1.03
3340	2	10	0.99	0.99
3340	10	10	0.99	0.97
3344		25	1.00	----
3344	10	50	1.00	----
3400		20	0.97	1.00
3400	1		0.96	0.99
3435		15	1.00	1.02
3435		45	0.99	1.00
3435	1	30	0.99	1.00
3459		35	1.01	1.01
3459	1		1.00	1.01
3459	1	45	1.01	1.03

## DISCUSSION

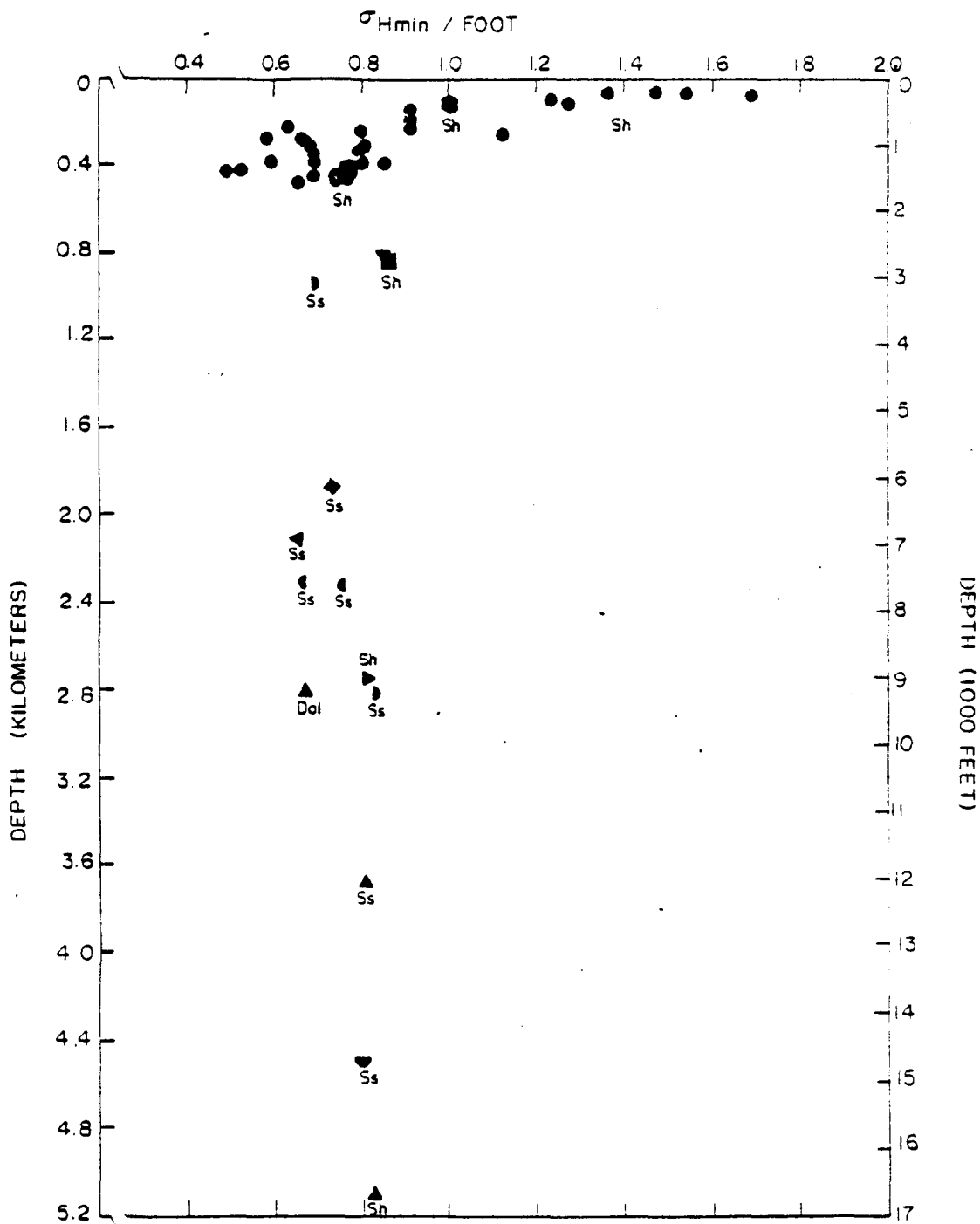
The subsurface stresses in sedimentary rocks are generally thought to increase linearly with depth: the vertical stress due to the overburden weight increases at about 1 psi per foot and the minimum-horizontal stress, on the average, increases at about 0.7 psi per foot (Figure 5).

The overburden stress-gradient of 1 psi per foot is generally accepted, although in geologically more recent environments (i.e., the Gulf Coast) this gradient is slightly less at 0.9 psi per foot. On the other hand, evidence is accumulating rapidly to suggest that the minimum-horizontal stress-gradient varies with rock type. For example, in hard rocks such as granites and quartzites the minimum stress-gradient can be as low as 0.5 psi per foot, in low-porosity sandstones it varies between 0.6 and 0.7 psi per foot and in weak shales and salt it can be as high as 0.9 psi per foot. It appears that the minimum stress-gradient decreases with increasing rock strength, but the underlying reason for this phenomenon is little understood.

The well-bore pressure gradient, required to induce and extend fractures in subsurface formations, is primarily dependent on the minimum stress-gradient and partially dependent on the overburden stress-gradient and formation pore-pressure gradient. It appears, therefore, that a good stimulation or fracture-treatment of a potential gas reservoir will, among other equally important variables, depend on the proper estimation or measurement of the stratigraphic distribution of the subsurface stress-gradients.

### Estimation of Devonian Shale Stress Gradients

Within the essentially monolithic Devonian Shale sequence, is it possible to determine whether or not significant variations in the minimum stress-



gradients exist among the major shale zones and to predict whether or not the fracture-treatment of prime shale zones will be successful? A partial answer to this question is suggested by the strain-relaxation tests on cored samples from three major shale zones in the Devonian Shale.

Among the many variables that affect strain-relaxation in rock samples, the two that follow are the most influential:

- Rock type; i.e., soft versus hard rock, and
- Magnitude of local, subsurface stresses.

All of the samples were obtained from the Devonian Shale which is finely stratified and almost entirely composed of micaceous minerals (Figure 6). The compositional differences between the major shale zones are, if any, very slight and do not significantly affect the strain-relaxation results.

The limited information that is available on the behavior of rock samples under load indicates that strain-relaxation rates increase markedly, but not necessarily linearly, when higher loads are removed from the samples. An important variable in this context is the magnitude of the maximum stress difference. In the subsurface, the maximum stress difference is that between the overburden stress and the minimum horizontal stress ( $\sigma_{O.B.} - \sigma_{Hmin}$ ).

The results of the strain-relaxation tests (Figure 4) show that samples from the Middle Brown Shale (MBs) zone relaxed at a rate 2 to 2.5 times greater than those representative of the upper Gray Shale (uGs) zones. Because the relaxation rates ( $\dot{\epsilon}$ ) are proportional to the local, subsurface, maximum stress differences ( $\sigma_{O.B.} - \sigma_{Hmin}$ ), we can write for each shale zone:

$$\begin{aligned} \dot{\epsilon}_{MBs} &= (\sigma_{O.B.} - \sigma_{Hmin}) \\ \text{and} & \\ \dot{\epsilon}_{uGs} &= (\sigma_{O.B.} - \sigma_{Hmin}) \end{aligned} \tag{1}$$



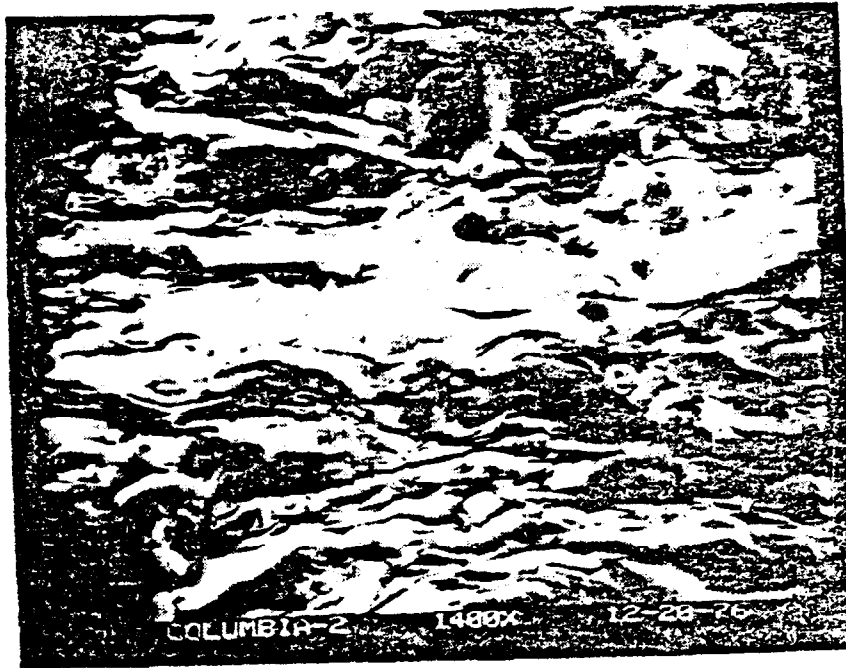


Figure 6. Scanning-electron micrograph of Middle Gray Shale (3056 ft) showing compacted clay particles. Scale: 1 cm = 20 microns.

The relaxation rates on the left of the proportionality can be written in terms of stress by multiplying  $\dot{\epsilon}$  by the Young's modulus (E) and an arbitrary time interval (t) and we get:

$$(E) (\dot{\epsilon}_{MBs})(t) = (\sigma_{O.B.} - \sigma_{Hmin}) \quad (2)$$

and

$$(E) (\dot{\epsilon}_{uGs})(t) = (\sigma_{O.B.} - \sigma_{Hmin})$$

Both sides of the proportionality are now in psi units, but the stresses on either side are not exactly equal to each other because the left-hand term refers to a stress in the sample and the right-hand term refers to the stress difference in the respective subsurface shale zones. We can now normalize proportionality (2) by dividing both sides by  $\sigma_{O.B.}$  or, what amounts to the

same thing since  $\sigma_{0.B.}$  increases by 1 psi/foot, by depth (feet):

$$(E)(\dot{\epsilon}_{MBs})(t)/ft = (1 - \sigma_{Hmin}/ft)$$

and

(3)

$$(E)(\dot{\epsilon}_{uGs})(t)/ft = (1 - \sigma_{Hmin}/ft)$$

The relaxation rates are now proportional to the minimum-horizontal stress-gradients in each of the shale zones. For the time being we assume that the Young's moduli (E) in both shale zones are equal to  $4 \times 10^6$  psi\* and that (t) is equal to 100 hours. The relaxation rates in the Middle Brown Shale zone and in the upper Gray Shale zones are  $4 \times 10^{-6}$ /hour and  $2 \times 10^{-6}$ /hour, respectively (Figure 4). Upon inserting and multiplying these values on the left side of proportionality (3), reversing sides and rearranging, we get:

$$(\sigma_{Hmin}/ft)_{MBs} = (1 - 1600/ft)$$

and

(4)

$$(\sigma_{Hmin}/ft)_{uGs} = (1 - 800/ft)$$

From well logs, the contact or boundary between the Gray Shale zone and the Middle Brown Shale zone in Columbia Well No. 20402 is at a depth of about 3340 feet. Upon dividing the numbers on the right side by 3340, we observe that the minimum stress-gradient in the Middle Brown Shale zone is proportional to  $(1 - 0.48) = 0.52$  psi/ft and in the overlying Gray Shale the minimum stress-gradient is proportional to  $(1 - 0.24) = 0.76$  psi/ft. The minimum stress-gradient in the Middle Brown Shale increases to 0.64 psi/ft if the Young's modulus (E) is reduced from  $4 \times 10^6$  to  $3 \times 10^6$  psi\* (Figure 7).

It should be understood, at this point, that the values of the minimum stress-gradients just obtained are only estimates; they do not equal the actual subsurface stress gradients. However, these estimates serve an

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\* The values for Young's modulus (E) were taken from Terra Tek Progress Report No. 3 and 4.

important purpose. They suggest that the minimum stress-gradient is higher (by as much as 30 percent) in the Gray Shales than in the Middle Brown Shale. This, in turn, suggests that an artificial fracture could be induced and extended at lower bottom-hole pressures in the Middle Brown Shale and, furthermore, if the bottom-hole treating-pressure can be kept below a peak-pressure level (1600 - 1700 psi at the well-head\*), the fracture propagating into the Middle Brown Shale can be contained and prevented from propagating upward into the Gray Shale zone because of the higher stress levels in this upper zone.

#### Comparison with Well-Bore Data

The discussion above may appear as so much conjecture. It is necessary, therefore, to compare the results and conclusions obtained from core-analysis (strain-relaxation) work with direct down-hole measurements.

The results of the Schlumberger Synergetic Log are abstracted in Figure 7A. It appears that the fracture-pressure-gradient profile of the Devonian Shale clearly identifies shale zones in which fractures can be induced and extended at lower bottom-hole pressures; i.e., the Lower Brown Shale (Marcellus) and the Middle Brown Shale. Although stresses do not appear explicitly in the calculation of the fracture-pressure-gradient profile, the log is an estimate of the minimum stress-gradients. The correlation between the subsurface data and the core-analysis (strain-relaxation) results is remarkably good (Figures 7A and B).

The result of the minifrac experiment performed by Terra Tek is also plotted in Figure 7A. The point represents the down-hole instantaneous shut-in pressure divided by the depth at which the measurement was made (2360 psi/2745 feet = 0.86). The agreement between the minifrac result

\*This upper limit in pressure is equal to the breakdown pressure plus the difference in the horizontal stresses between the upper Gray Shales and the Middle Brown Shale.

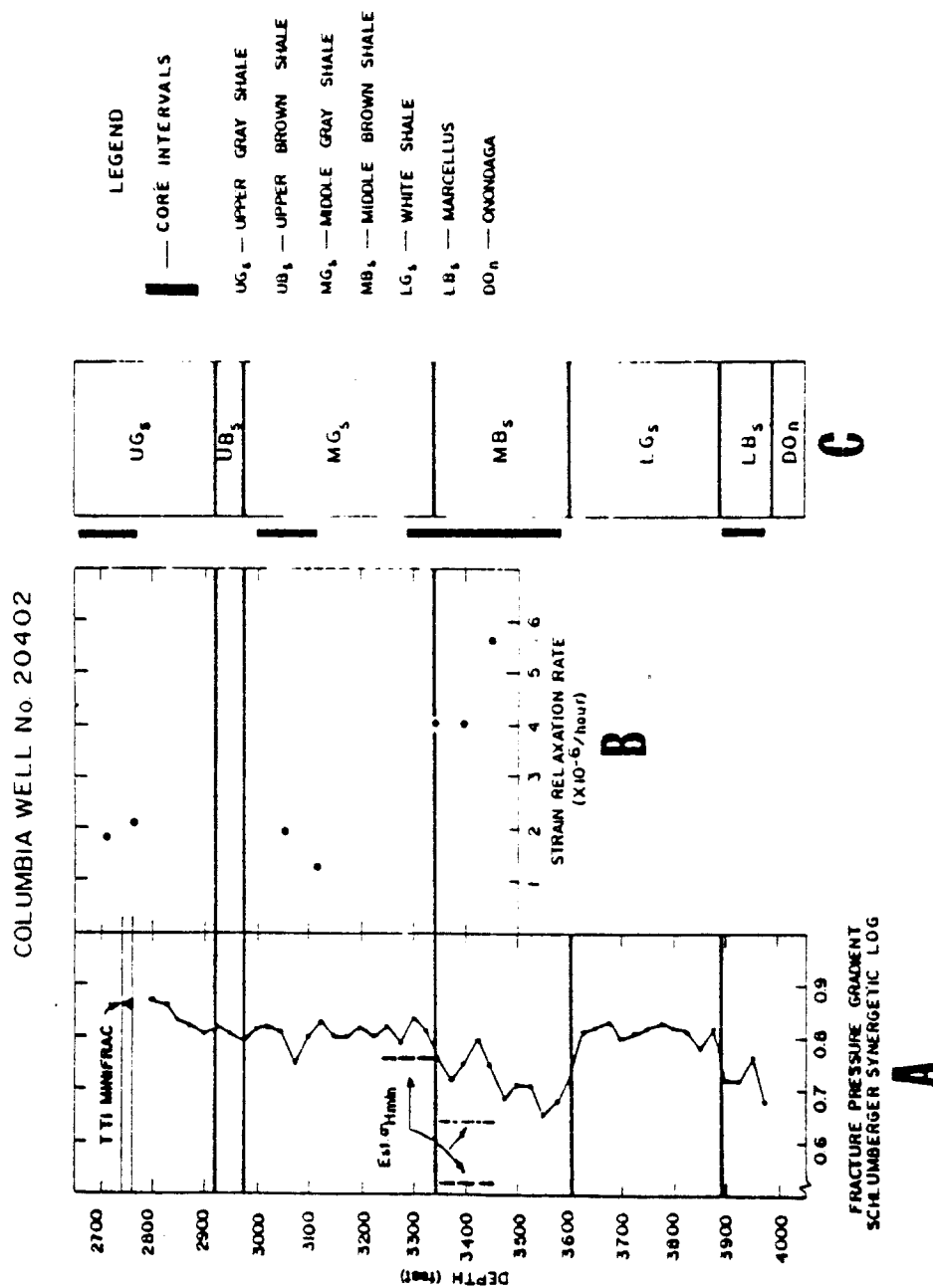


Figure 7. A. Correlation between the minimum-horizontal-stress estimates (--- for  $E = 3 \times 10^6$  psi and ---- for  $E = 4 \times 10^6$  psi) and the fracture-pressure-gradient profile and the instantaneous shut-in pressure gradient determined down-hole.  
 B. Same as Figure 25.  
 C. Stratigraphic column of Devonian Shale in Columbia Well No. 20402.

and the Synergetic Log which starts just below the minifrac interval is quite good. It confirms the inference that high subsurface stress-gradients are proportional to low strain-relaxation rates in cored samples.

RECOMMENDATIONS  
FOR COLUMBIA WELL NO. 20402

1. The prime, gas-bearing, candidate shale zones for fracture (MHF) treatment in Columbia Well No. 20402 which is presently shut-in are:  
    The Middle Brown Shale and  
    The Marcellus.
2. Fracture treatment should be confined to these two shales only, if the object is to contain the induced fractures within these shale zones and increase gas production.
3. The higher stresses in the Gray Shales and White Slate will aid in the containment of the induced fracture within the Middle Brown Shale and in the Marcellus, prevent them from propagating either upward or downward into the barrier zones, and promote the lateral extension of the fracture.
4. The bottom-hole treating-pressure (BHTP) should not be greater by 400 to 800 psi above the bottom-hole breakdown pressure, which is about 800 to 900 psi at the well-head for the Middle Brown Shale. If BHTP greatly exceeds this peak pressure the barrier zones (Gray Shale and White Slate) may breakdown as well.
5. These recommendations, based only on the strain-relaxation tests and a single minifrac test, should be carefully checked against previous MHF experience in Wells No. 20401 and 20403, in which the potential barrier zones were perforated and fractured.
6. Unsolved, but pertinent problems remaining are, among others:
  - Exact fracture-density variation among zones
  - Secondary-porosity prediction
  - Prediction of formation breakdown-pressure.

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APPENDIX I  
DATA ANALYSIS FOR  
STRAIN ROSETTES

The stress in a solid material (rock) subjected to uniaxial stress can be determined experimentally by attaching a strain gage oriented in the direction of the applied stress. The stress ( $S$ ) is then computed, in terms of the measured strain ( $\epsilon$ ), from  $S = E\epsilon$ , where  $E$  is the elastic Young's Modulus of the material. The strain is generally small (a few parts in 1000000); hence sensitive instruments are required for measuring it. Originally, strain gages were mechanical or optical, but these have now been almost completely replaced by electrical gages. This type of gage contains a wire element whose electrical resistance varies with its deformation. The gage is cemented to the test specimen, the strain in the specimen being measured as a function of the change in the electrical resistance of the wire element.

A single strain gage oriented in the direction of a uniaxially applied stress is sufficient for computing the stress in the test specimen. For biaxial stress, one might suppose that two strain gages would be sufficient; this would be true if the directions of the principal stresses were known, but this is not usually the case. To determine the direction of the principal stresses in addition to their magnitudes, three values of strain are required. As a matter of practical convenience, the linear strains are obtained by using a combination of three resistance gages: three gages set with their axes at  $45^\circ$  to each other (central figure in Figure 8). This combination is known as a strain rosette. The three gages are electrically insulated from each other and are used to determine the strain at the surface of a specimen to which they are attached.

For a 45° strain rosette, the principal strains are

$$\epsilon_{\max} = \frac{\epsilon_a + \epsilon_c}{2} + 0.71 \left[ (\epsilon_a - \epsilon_b)^2 + (\epsilon_b - \epsilon_c)^2 \right]^{1/2}$$

$$\epsilon_{\min}$$

and the direction of the maximum principal strain is defined by

$$\tan 2\theta = \frac{\epsilon_a + \epsilon_c - 2\epsilon_b}{\epsilon_a - \epsilon_c}$$

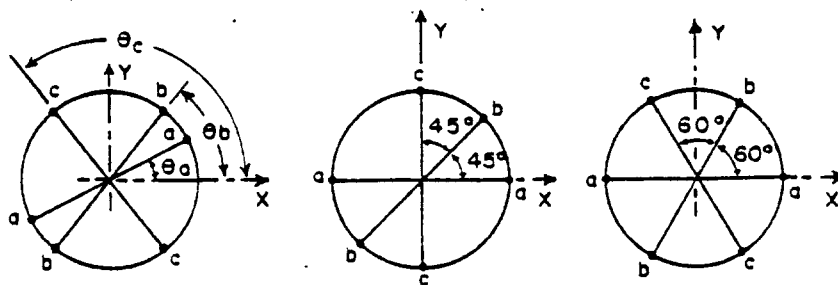


Figure 8. Strain Rosettes.

APPENDIX II  
STRAIN TIME PLOTS.

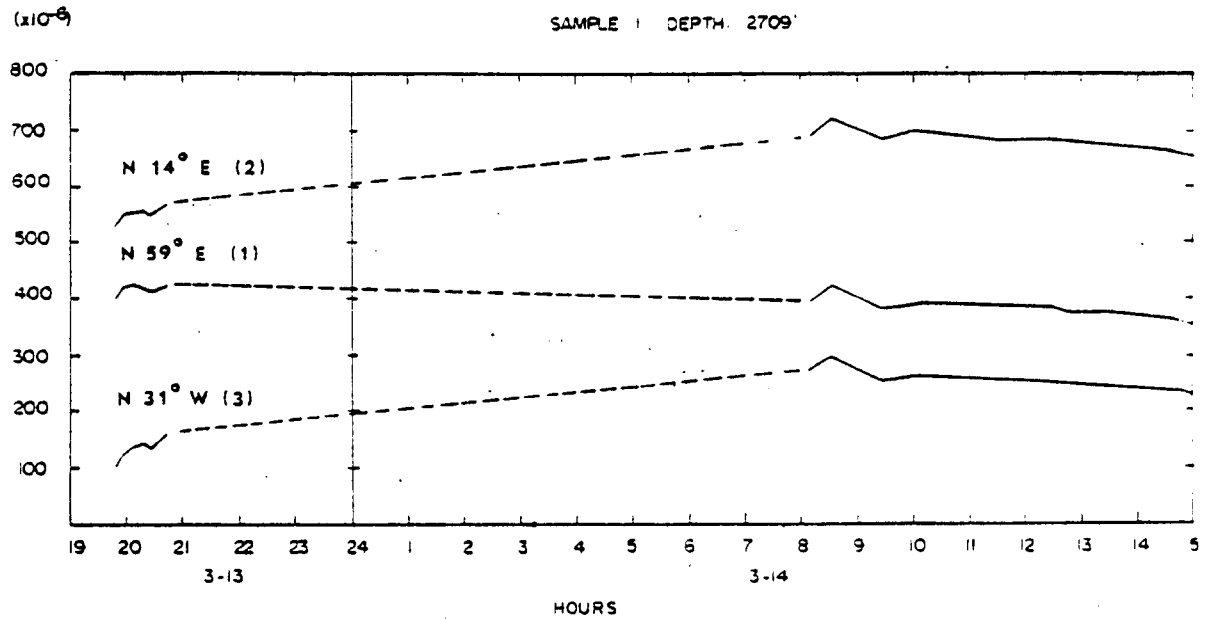


Figure A1. Strain-relaxation-time plots of Devonian Shale.

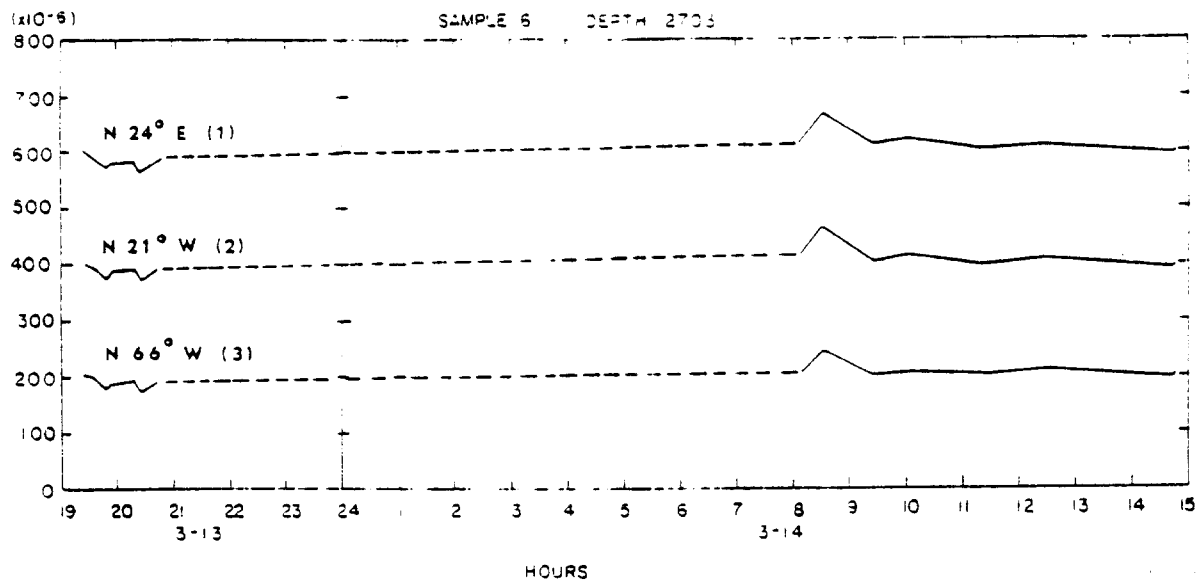


Figure A2. Strain-relaxation-time plots of Devonian Shale.

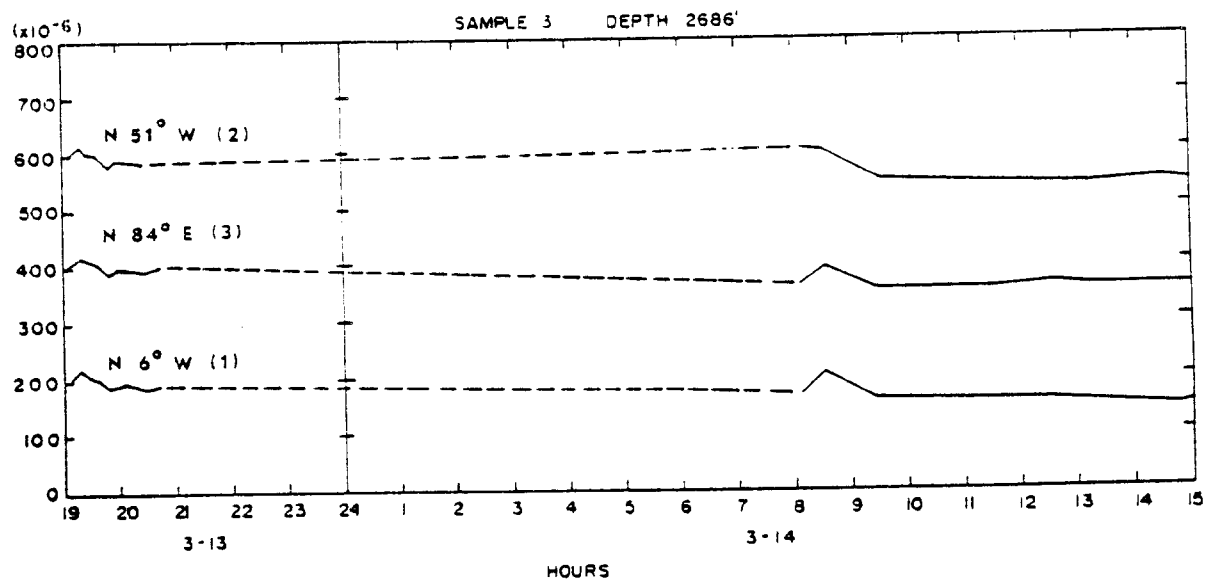


Figure A3. Strain-relaxation-time plots of Devonian Shale.

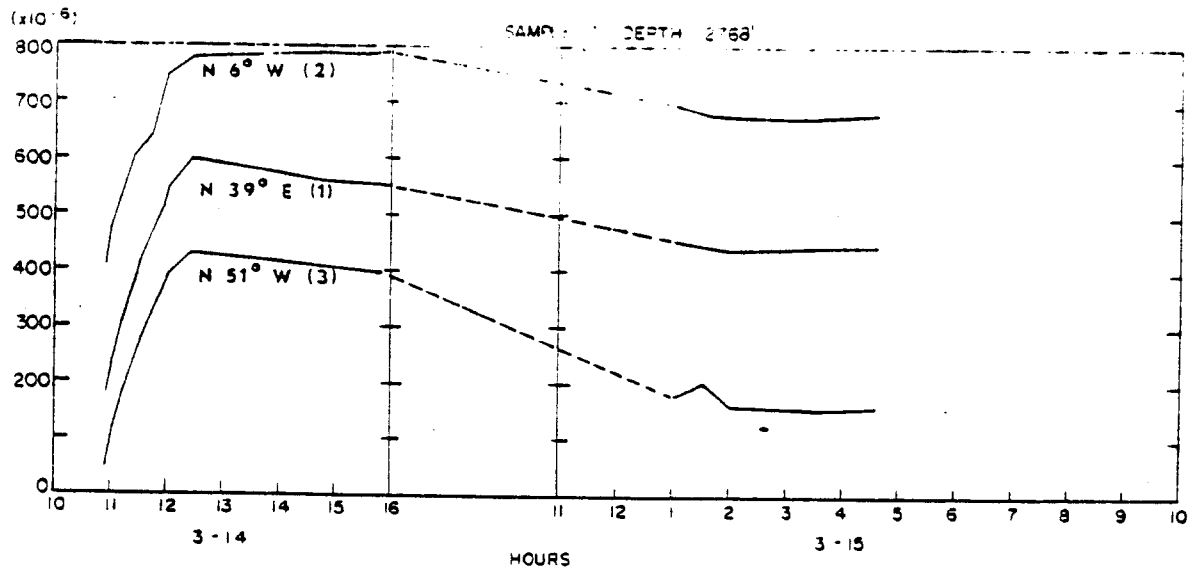


Figure A4. Strain-relaxation-time plots of Devonian Shale.

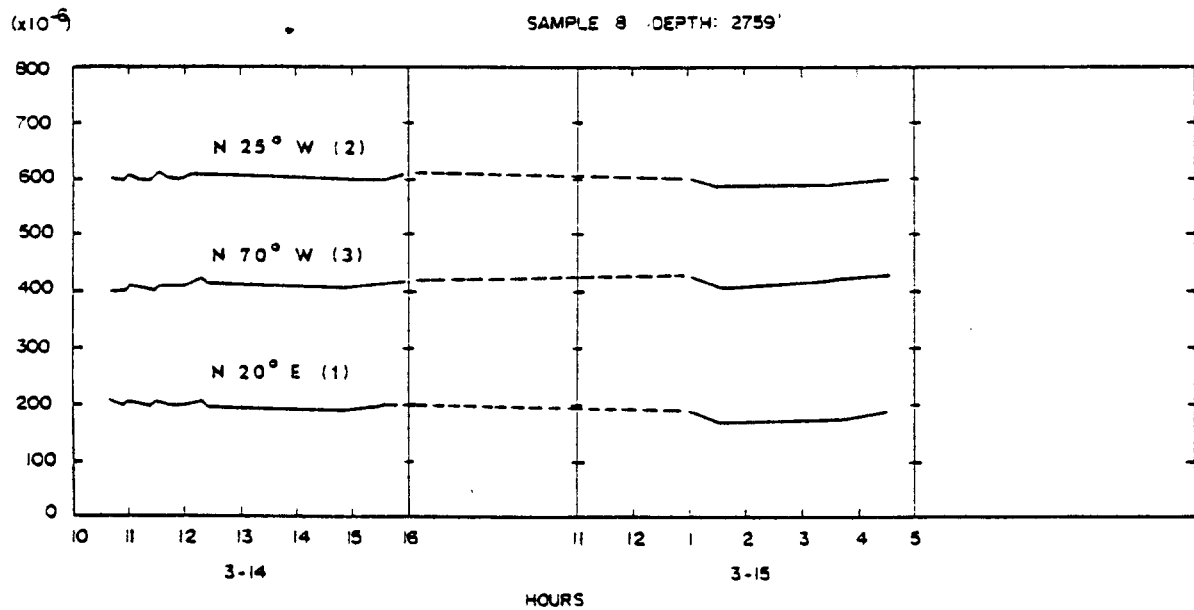


Figure A5. Strain-relaxation-time plots of Devonian Shale.

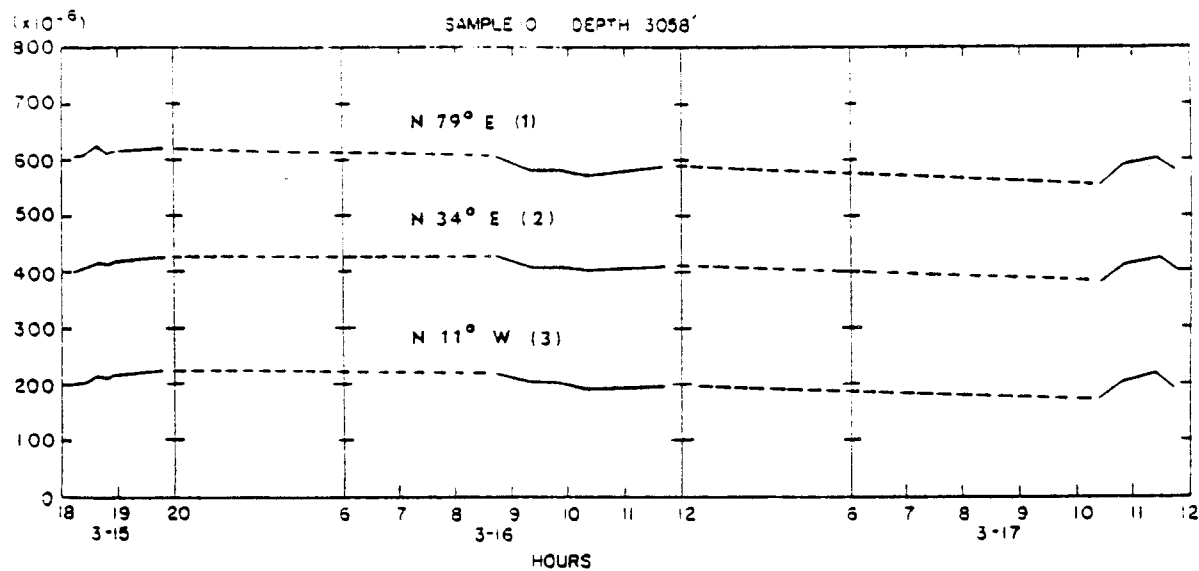


Figure A6. Strain-relaxation-time plots of Devonian Shale.

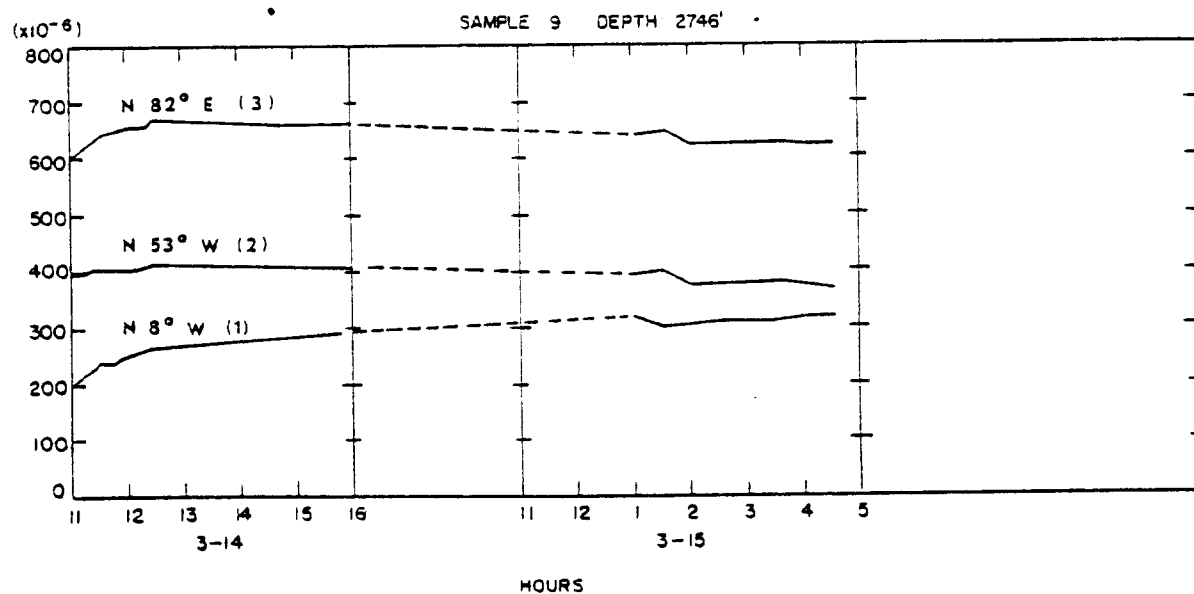


Figure A7. Strain-relaxation-time plots of Devonian Shale.



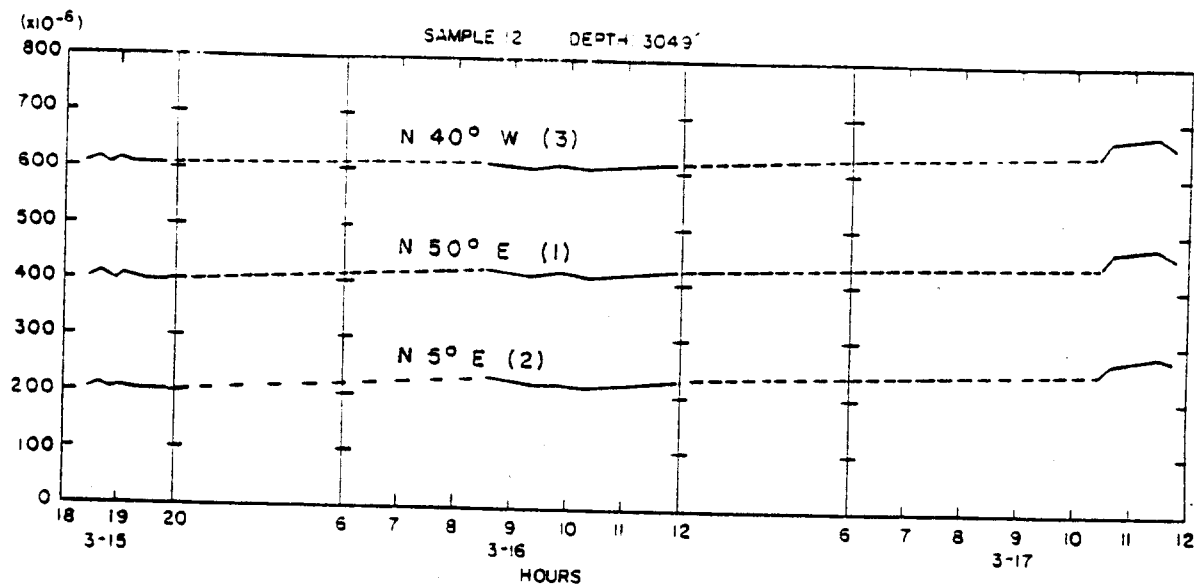


Figure A8. Strain-relaxation-time plots of Devonian Shale.

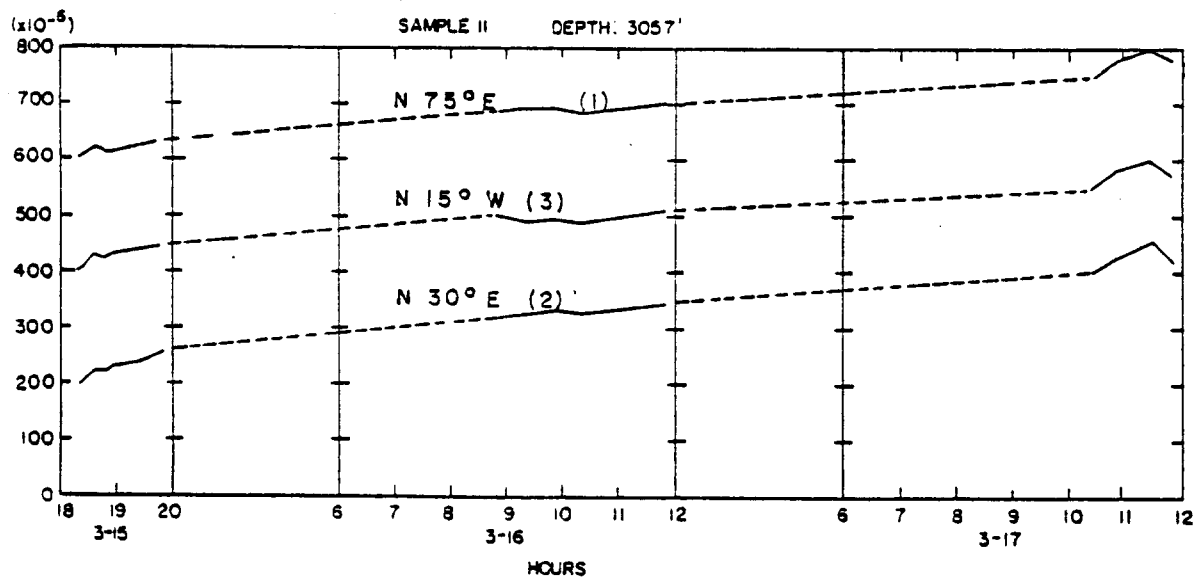


Figure A9. Strain-relaxation-time plots of Devonian Shale.

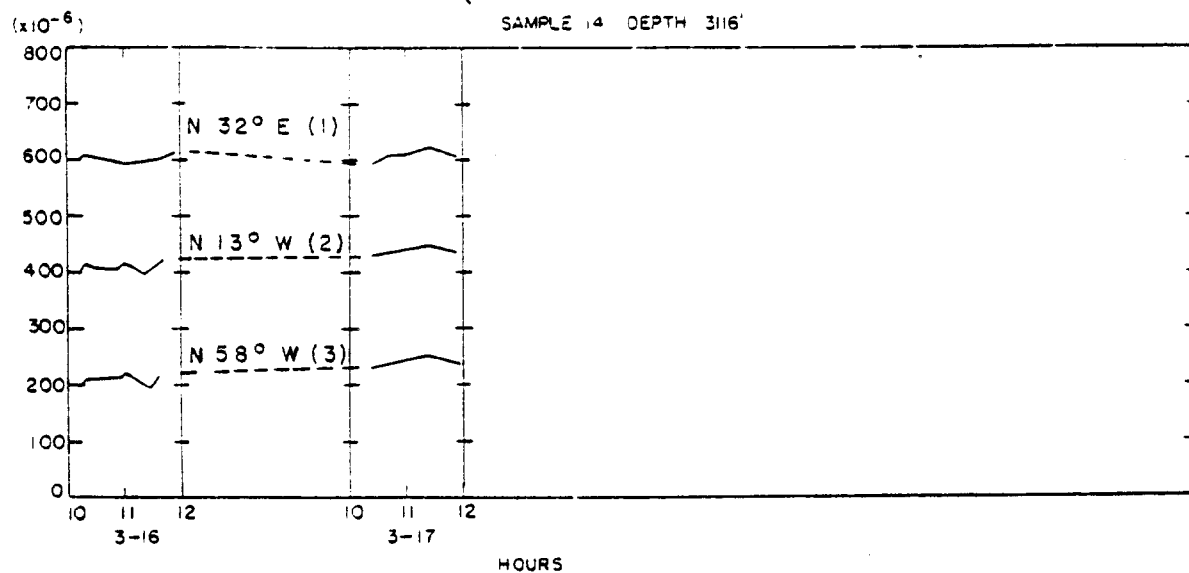


Figure A10. Strain-relaxation-time plots of Devonian Shale.

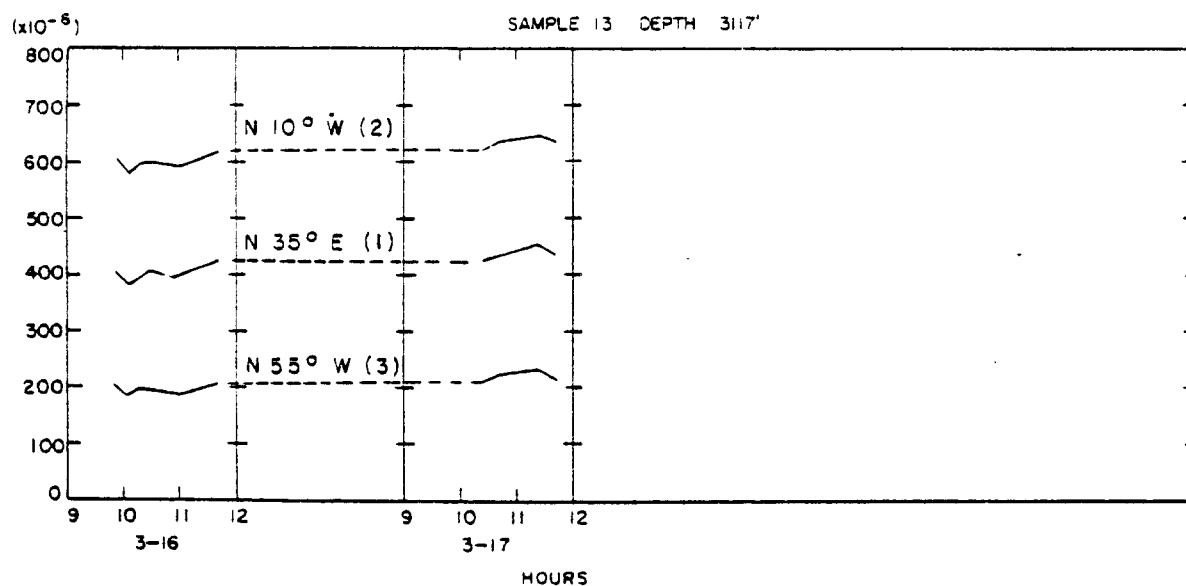


Figure A11. Strain-relaxation-time plots of Devonian Shale.

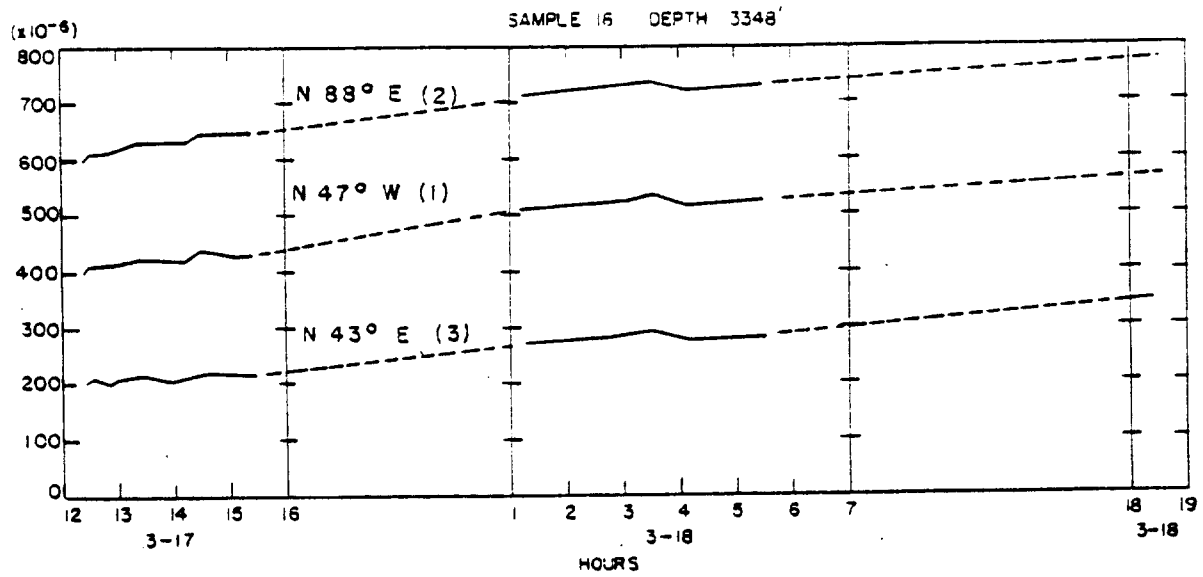


Figure A12. Strain-relaxation-time plots of Devonian Shale.

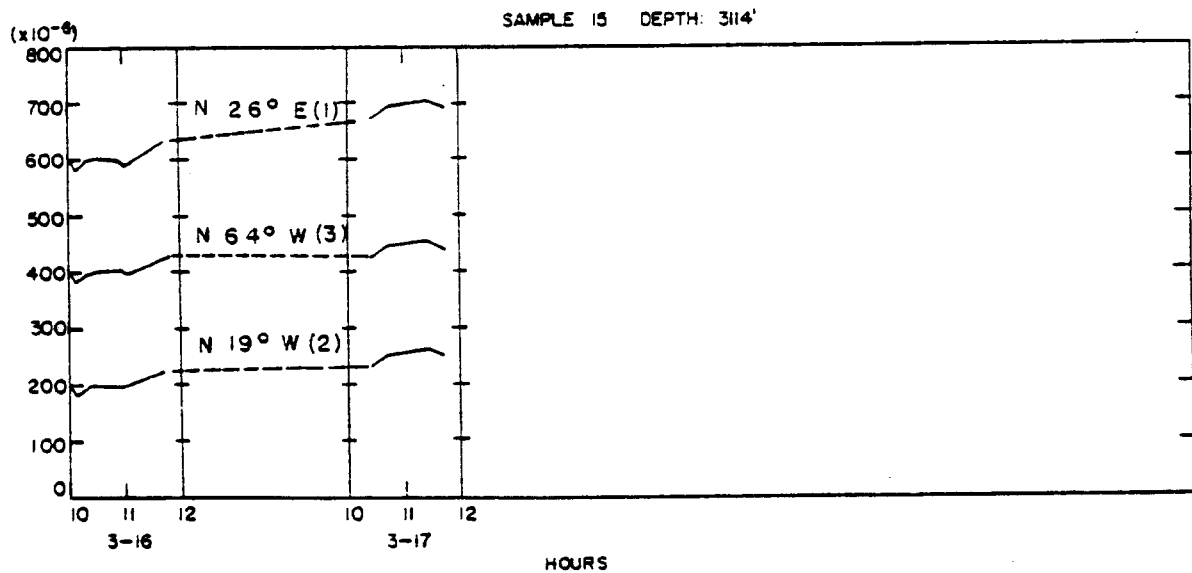


Figure A13. Strain-relaxation-time plots of Devonian Shale.

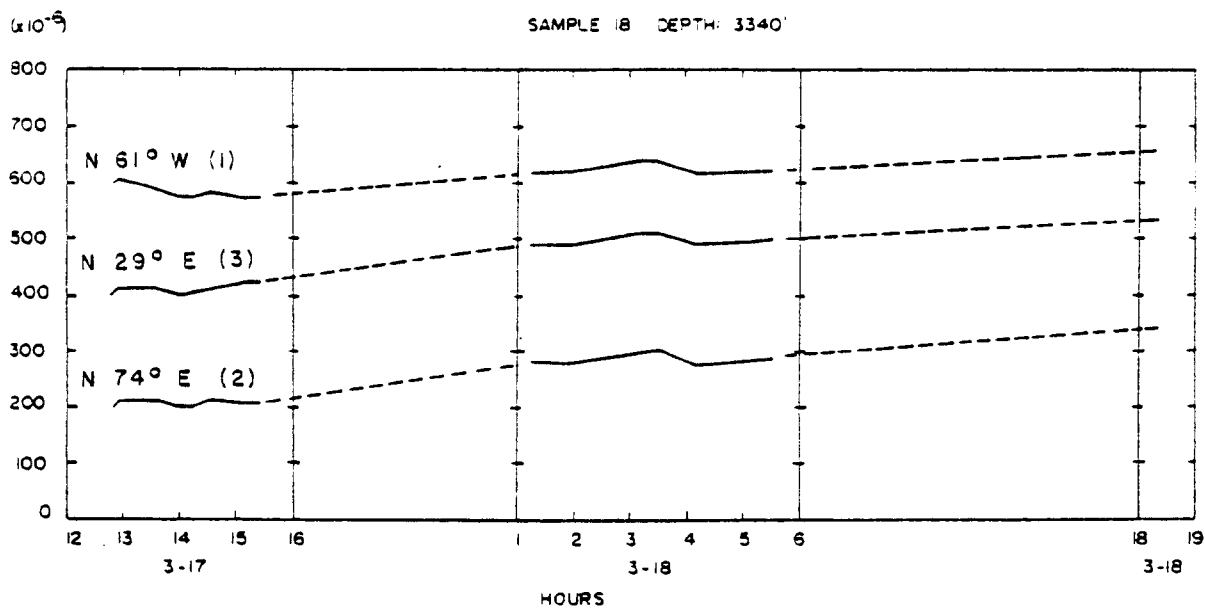


Figure A14. Strain-relaxation-time plots of Devonian Shale.

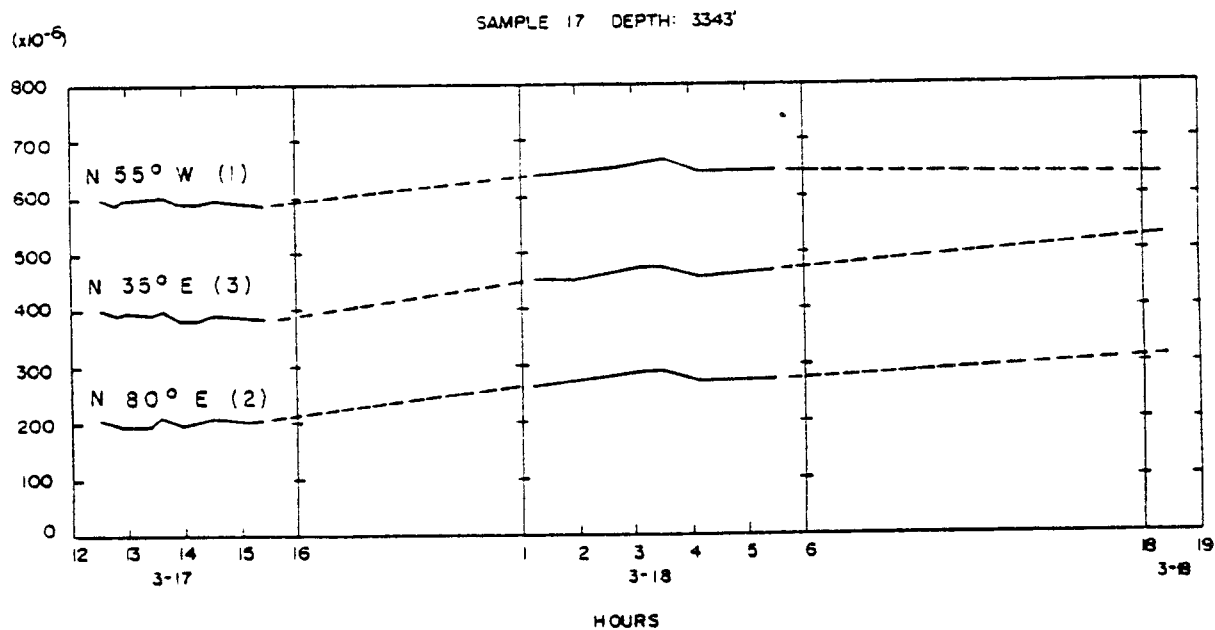


Figure A15. Strain-relaxation-time plots of Devonian Shale.

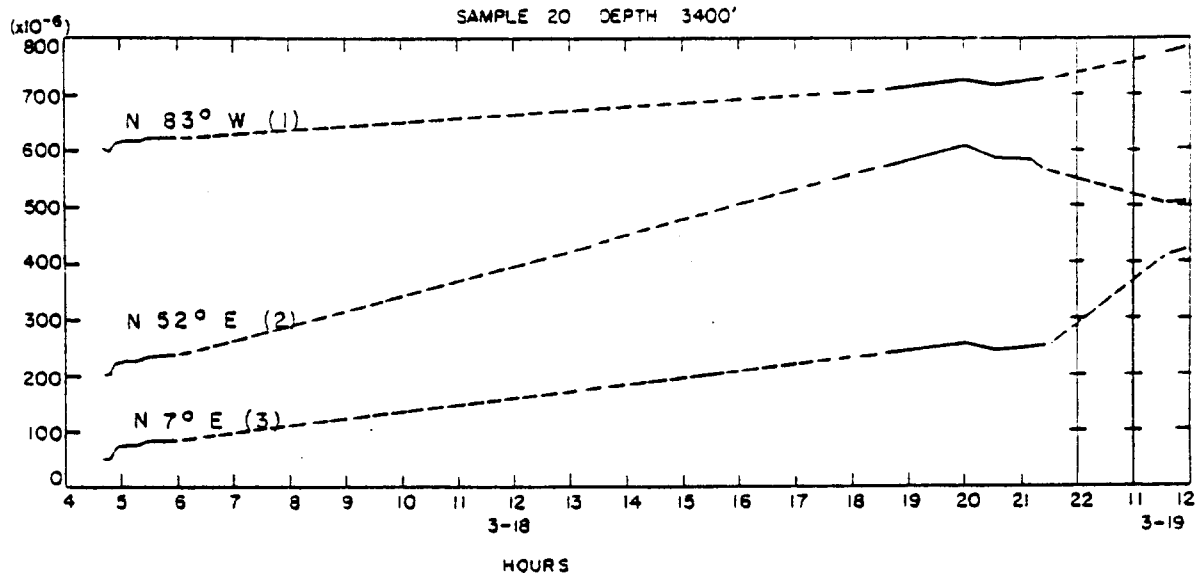


Figure A16. Strain-relaxation-time plots of Devonian Shale.

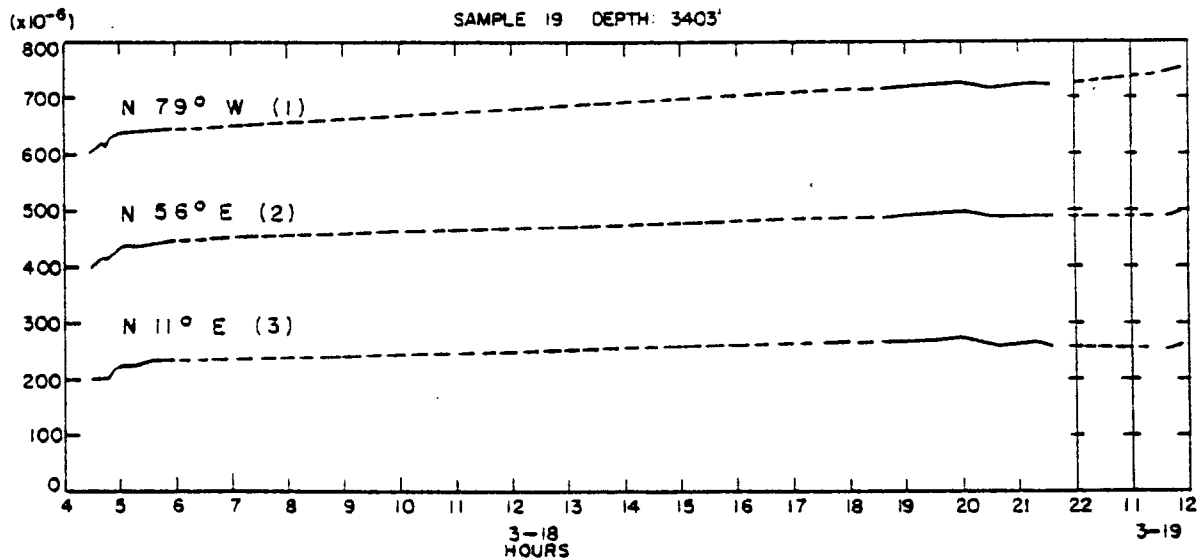


Figure A17. Strain-relaxation-time plots of Devonian Shale.

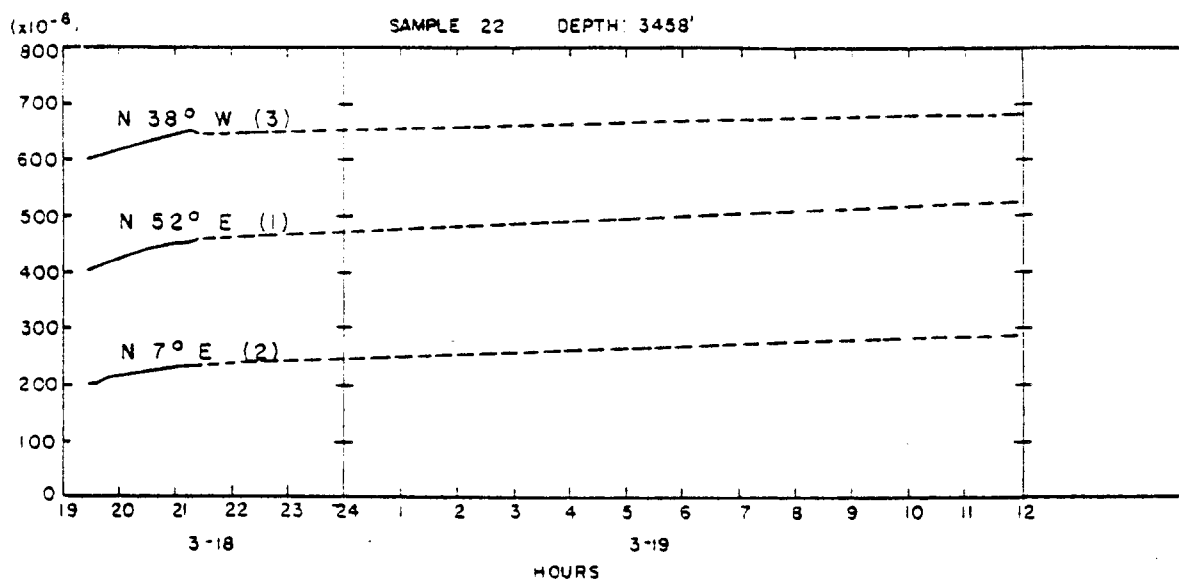


Figure A18. Strain-relaxation-time plots of Devonian Shale.

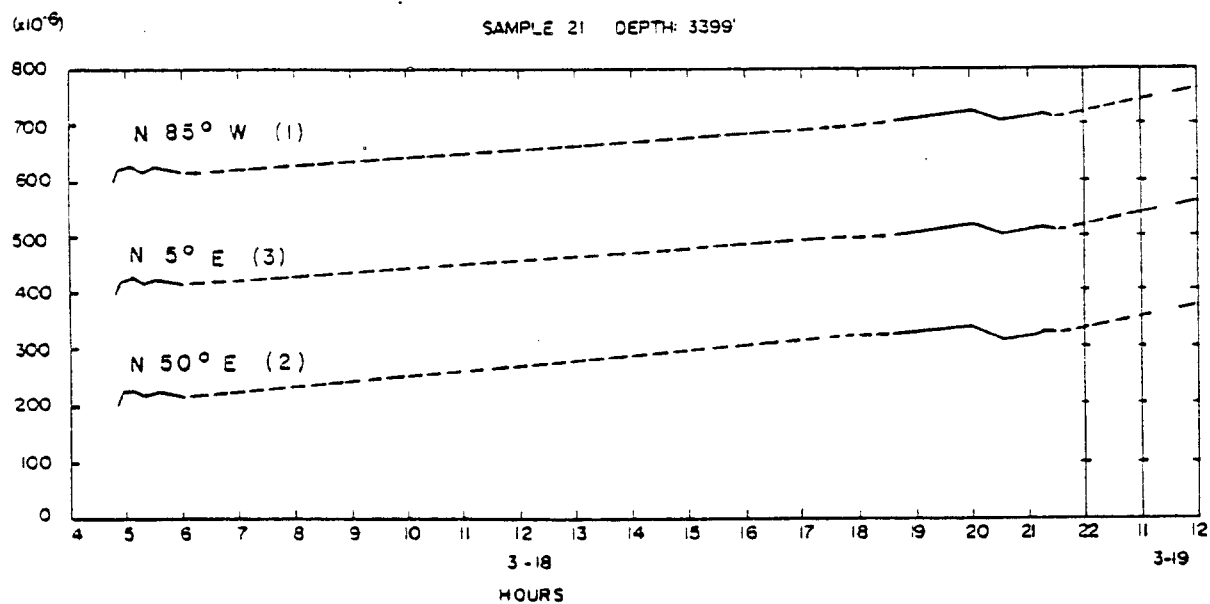


Figure A19. Strain-relaxation-time plots of Devonian Shale.

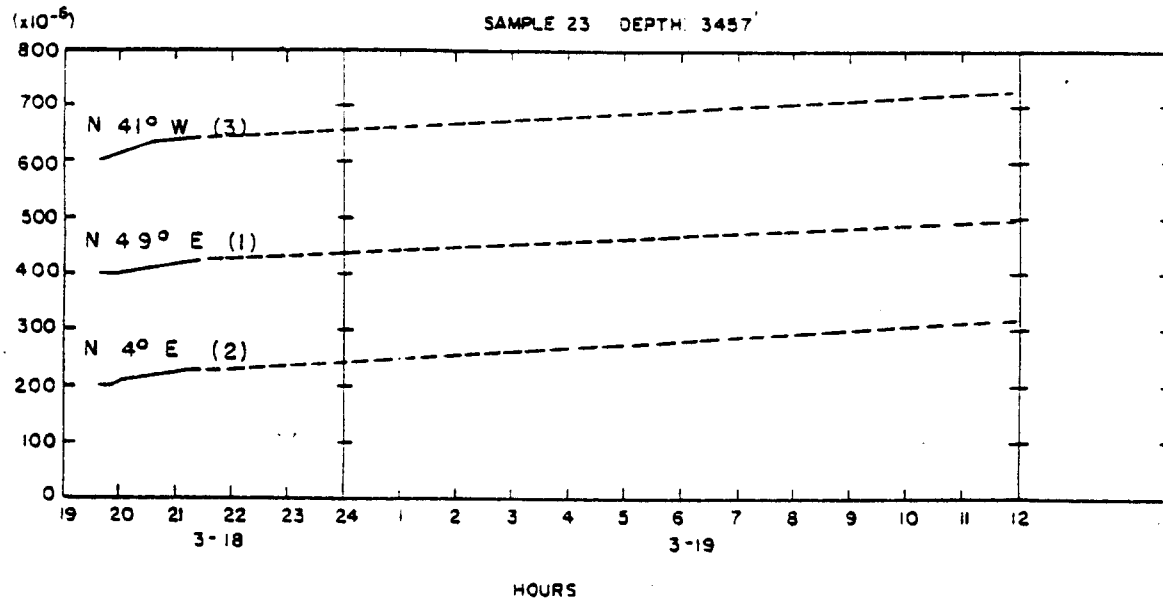


Figure A20. Strain-relaxation-time plots of Devonian Shale.

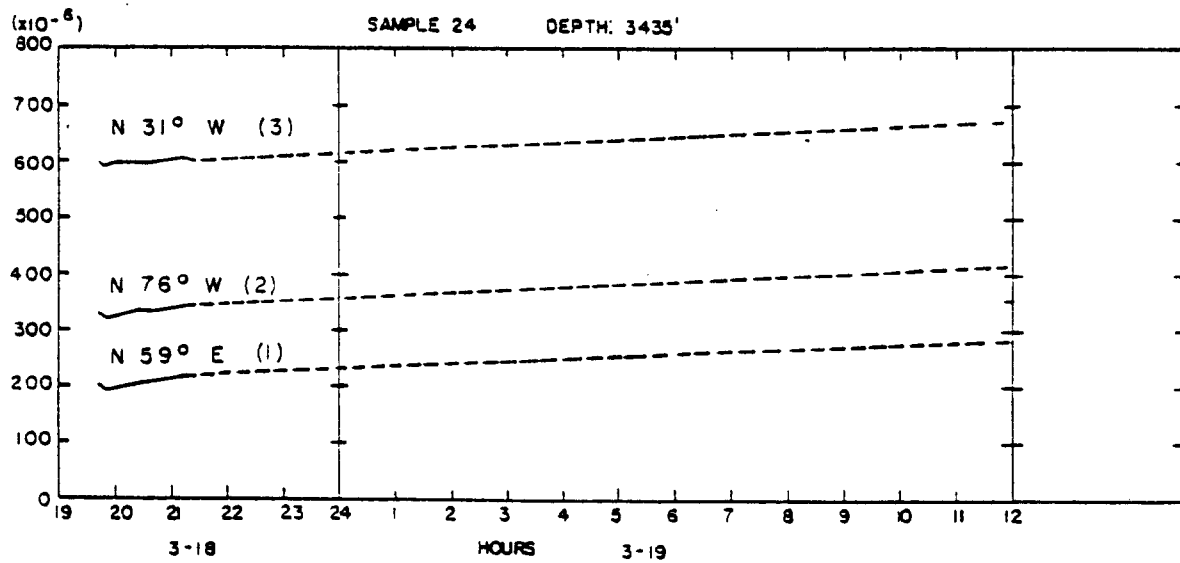


Figure A21. Strain-relaxation-time plots of Devonian Shale.

TERRA TEK REPORT

ROCK MECHANICS STUDIES RELATED TO  
MHF OF EASTERN UNITED STATES DEVONIAN SHALES



ROCK MECHANICS STUDIES RELATED TO MASSIVE HYDRAULIC FRACTURING  
OF EASTERN UNITED STATES DEVONIAN SHALES  
FINAL CORE ANALYSIS REPORT

WORK COMPLETED FOR  
COLUMBIA GAS SYSTEM SERVICE CORPORATION  
UNDER PRIME CONTRACT NO. E(46-1)-8014

By .

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TR 77-16  
March 1977

## SUMMARY

Ultrasonic velocities and elastic moduli were determined for cores taken from several layers in wells #20402 and #20403. The effect of *in situ* conditions (pressure and temperature) on the measured velocities was very small; a change on the order of 2 percent or less for pressures ranging from atmospheric to 4,000 psi and temperatures up to 100°F. Laboratory-measured densities and p-wave velocities are in good agreement with log-measured values. However, the log-measured shear wave velocities had to be interpreted as surface wave velocities in order to obtain good agreement with the corresponding laboratory-measured values. Dynamic Young's moduli have been calculated using the log-derived bulk densities, p-wave and s-wave (after correction) velocities. The static moduli were determined from triaxial compression tests on the Upper Brown and Lower Gray Shales. The static moduli are lower than the dynamic moduli; for the samples tested, the difference is 20 to 30 percent.

Critical stress-intensity factors were determined experimentally for the gray shales. Permeability of shale samples as well as a flow conductivity through propped and unpropped fractured specimens have been determined under simulated downhole stress conditions. Both the permeability and the unpropped-fracture conductivity are strongly affected by the confining pressure. On the other hand, conductivity of the propped fracture shows insignificant reduction with application of confining pressure to simulated *in situ* conditions.

## RECOMMENDATIONS

Fracture containment analysis indicates that the gray shales and Onondaga Limestone will act as barriers to fracture growth in the brown shales. The Onondaga Limestone forms a strong barrier to fractures in the Lower Brown Shale. The gray shales are not strong barriers, however, and can also be broken down if fluid pressure is too high. The brown shales will not form barriers to fractures in the gray shales. The following are recommendations to optimize MHF treatments in this test area.

1. For the Upper Brown Shale, the Upper Gray and Middle Gray Shales will form weak barriers and a treatment pressure within the formation (FTP) of no more than 200 psi in excess of the minimum horizontal stress for the Upper Brown Shale, plus any expected drop in pressure across the perforations, will be required for fracture containment.

2. In treating the Middle Brown Shale, the fracture will propagate into both the Middle Gray and Lower Gray Shales. There is the possibility that later fractures initiated in the Upper Brown and Lower Brown Shales will interact with the earlier fracture created in the Middle Brown Shale. Therefore, in a multi-stage MHF it is important that the Middle Brown Shale formation, having the best gas potential, should be fractured first.

3. Penetration of a fracture which is initiated in the Middle Brown Shale into the Middle Gray Shale can be minimized to 50 feet or less by maintaining the FTP during hydraulic fracturing of the Middle Brown Shale layer within 400 psi in excess of the minimum horizontal stress in that zone, plus any expected drop in pressure across the perforations.

4. Corresponding considerations for the Lower Brown and Lower Gray Shales leads to a FTP of at most 250 psi in excess of the minimum horizontal stress for the Lower Brown Shale, plus any expected drop in pressure across the perforations.

## ABSTRACT

Laboratory core analyses were performed on cores taken from Columbia Gas System Service Corporation wells #20402 and #20403, located in Lincoln County, West Virginia. Physical and mechanical properties, determined at simulated *in situ* conditions (where applicable), were compared directly with the Birdwell (3-D) and Schlumberger logs. Densities compare favorably; compressional and shear wave comparisons are complicated by anisotropic material behavior and discrepancy in field-measured shear wave velocities. Good correlation is obtained for compression and shear wave velocities parallel to the wellbore provided the log-determined shear wave velocity is assumed to be the surface wave velocity. Measured physical and mechanical properties are combined with theoretical models to assess possible fracture containment. The analysis indicates that for successful massive hydraulic fracture in the Middle Brown Shale the formation treatment pressure (FTP) to extend the fracture should not be greater than 400 psi above the minimum *in situ* stress in order to prevent deep excursion of the fracture into the barrier layers. For fracturing treatment of the Upper Brown Shale and the Lower Brown Shale, the corresponding treating pressure should be limited to 200 psi and 250 psi, respectively, in excess of the minimum horizontal stress in the respective zone. The Middle Brown Shale should be the first formation treated. Fracture conductivities at simulated *in situ* conditions for an unpropped fracture and for fractures propped with 0.56 lb/ft<sup>2</sup> of 100 mesh and 1.44 lb/ft<sup>2</sup> of 20/40 sand were found to be 1 md-cm, 1100 md-cm and 8000 md-cm, respectively, for the Lower Brown Shale.

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APPENDIX IV: REPORTS ON ROCK MECHANICS  
RESEARCH BY TERRA TEK

## BACKGROUND

Under contract E(46-1)-8014 from the Energy Research and Development Administration, Columbia Gas System Service Corporation was to:

1. Determine the technical and economic feasibility of hydraulic fracturing of the Devonian shales, and
2. Identify the fracture distribution, gas distribution, and ways to achieve economical production by application of stimulation research findings.

Three wells were drilled in Lincoln County, West Virginia. Different fracture treatments were planned for each well to optimize the stimulation treatment. Terra Tek, under subcontract from Columbia Gas System Service Corporation, was to determine the physical characteristics of the formations to better design and interpret the results of the massive hydraulic fracture (MHF) tests. This work was performed through three tasks:

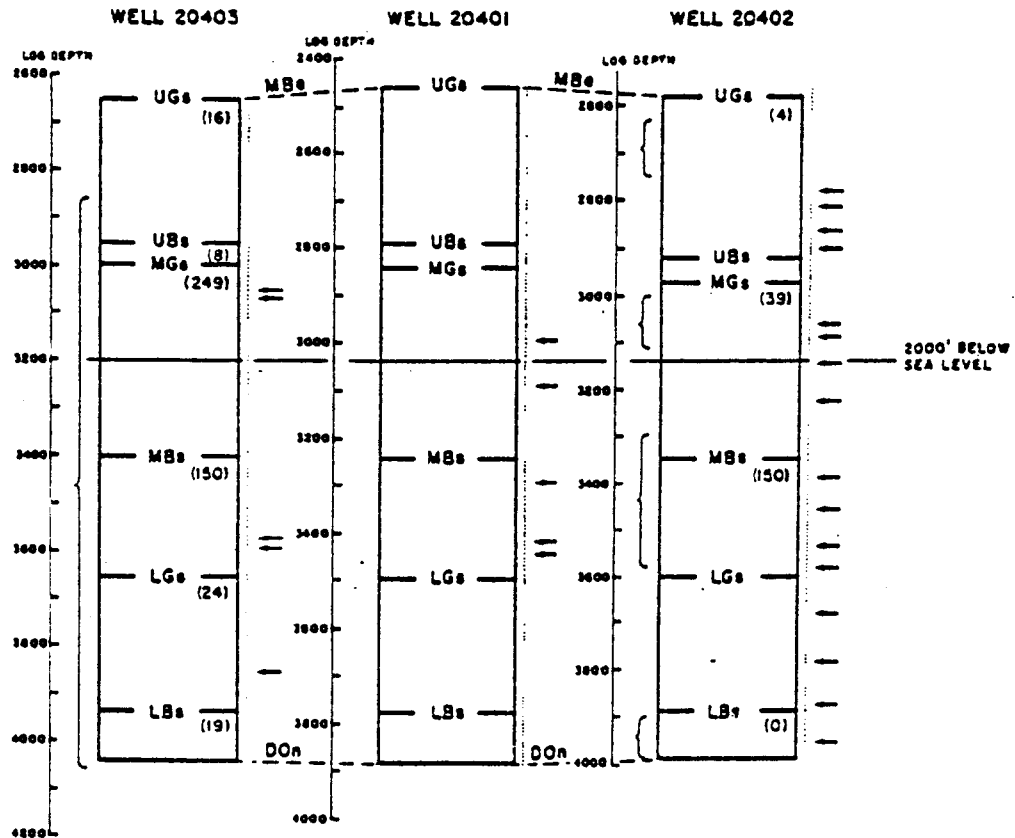
1. Core Analysis: Evaluate physical and mechanical properties to determine fracture containment and pumping schedule.
2. Strain Relaxation on Recovered Cores: Estimate the *in situ* principal stresses in principal shale zones. Results were to be used to determine fracture containment, pumping shedule and fracture direction.
3. Open Hole Tests: Determine *in situ* stresses and their directions.

The present report describes the results and implications to massive hydraulic fracture design from core analyses.

Laboratory test data gathered in support of log interpretation, MHF design and reservoir engineering are briefly summarized. This consists of physical and mechanical properties data on cores retrieved from wells #20402 and #20403. Since the subsurface bedding is very nearly horizontal in the area of these wells and there is no major tectonic displacements between them, the shale layers are expected to be similar in the three wells, #20401, #20402 and #20403. Figure 1 is a schematic that summarizes all geological findings for the wells.

Except for density determination, all tests were performed at simulated "down-hole" pressure conditions. These include sonic wave velocities, triaxial extension and compression, fracture toughness, rock permeability and fracture permeability for the propped and unpropped conditions. The test data is then used to assess fracture containment. A fracturing analysis is presented along with a recommended treatment pressure limitations.

## SUMMARY OF GEOLOGICAL FINDINGS (COLUMBIA MHF WELLS, Lincoln Co., W. Va.)



### LEGEND

- |   |  |
|---|--|
| MB <sub>s</sub> - BASE OF BEEBA SANDSTONE               | LB <sub>s</sub> - TOP OF LOWER BROWN SHALE                 |
| UG <sub>s</sub> - TOP OF UPPER GRAY SHALE AND SILTSTONE | DON - TOP OF CORNIFEROUS (ONONDAGA) LIMESTONE              |
| UB <sub>s</sub> - TOP OF UPPER BROWN SHALE              | — - GAS SHOW FROM SIBILATION LOG                           |
| MG <sub>s</sub> - TOP OF MIDDLE GRAY SHALE              | - PROBABLE PERMEABLE INTERVAL FROM RESISTIVITY CROSS PLOTS |
| MB <sub>s</sub> - TOP OF MIDDLE BROWN SHALE             | (5) - FEET OF OBSERVED VERTICAL FRACTURES                  |
| LG <sub>s</sub> - TOP OF LOWER GRAY SHALE               | { - CORED INTERVAL   |

Figure 1. Summary of geological findings (from Ranostaj, 1976).

## LABORATORY MEASUREMENTS OF PHYSICAL PROPERTIES AND ULTRASONIC VELOCITIES

Physical properties and ultrasonic velocities have been determined in the laboratory on cores recovered from wells #20402 and #20403. Bulk dry density, grain density and porosity measurements have been carried out on samples obtained from depths of 2980, 3446, 3760 and 3996 feet. Three specimens from each depth have been used for measurement. To determine the bulk dry density, the samples are dried at 110°F, weighed, then the total volume is measured in a mercury porometer. Grain density is measured using crushed or pulverized specimen. The percent porosity is determined from the bulk dry density,  $\rho_D$ , and the grain density,  $\rho_g$ , by the following equation

$$\text{Percent Porosity} = 100 \left( 1 - \frac{\rho_D}{\rho_g} \right)$$

Results of the physical property measurements are given in Table I.

Both the compression wave and shear wave velocities were measured on samples subjected to simulated *in situ* conditions. Measurements were performed both in the axial (normal to bedding) and the horizontal (parallel to bedding) directions. This data, along with the measured bulk density, has been used to calculate the dynamic elastic moduli of the materials. The measured moduli were used to establish the validity of comparable data obtained for sonic logs. The comparison of log-derived and laboratory-measured data is illustrated in Figures 2 and 3, and in further detail in Appendices A and B.

The findings are summarized as follows:

1. The laboratory bulk density measurements show good agreement with the wireline log measurements.
2. The laboratory compression wave velocities measured in the vertical direction are in good agreement with the wireline log data.

3. The laboratory shear wave velocity measurements, taken in the same direction as the log measurements, are consistently higher.

4. Laboratory-measured Poisson's ratios in the different directions are only slightly different (same ratio of P-wave to S-wave velocities).

5. Young's modulus and the bulk modulus in the horizontal plane, derived assuming isotropy of the Devonian shales, are vastly different from those moduli in the vertical direction.

Since the elastic properties in the horizontal plane are pertinent to containment analysis of massive hydraulic fractures and not the vertical velocities usually measured by a wireline sonic log, care should be exercised when interpreting the log data for containment analysis. When the horizontal and vertical velocities are different, the most that can be obtained from a wireline elastic properties log is a close approximation of Poisson's ratios. This approximation is only possible if the ratio of the compression wave velocity to the shear wave velocity remains nearly constant in different directions, as is the case for the Devonian shales.

At the present time, there is an unaccountable discrepancy between the laboratory-determined and log-measured shear wave velocities. A possible explanation for the discrepancy is that the wireline logs were detecting the arrival of a surface wave instead of the shear wave. If the log shear wave velocities are recalculated based on the assumption that the log-measured wave was really a surface wave, then (see Figure 4) the corrected log velocities (denoted by ●) are in very good agreement with the laboratory data (denoted by x). This is discussed in more detail in Appendix A.

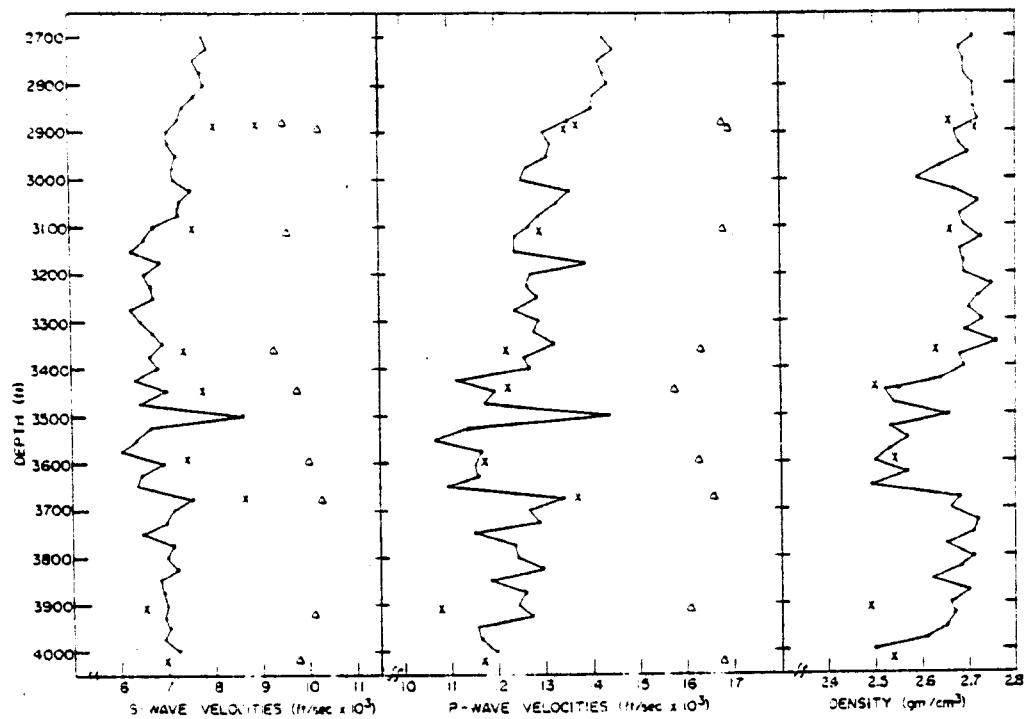


Figure 2. Comparison of laboratory data (Terra Tek, Inc.) with data from a Schlumberger experimental s-wave logging tool on Columbia Gas System Service Corporation well #20403. "X" indicates sample orientation parallel to borehole, "Δ" - perpendicular.

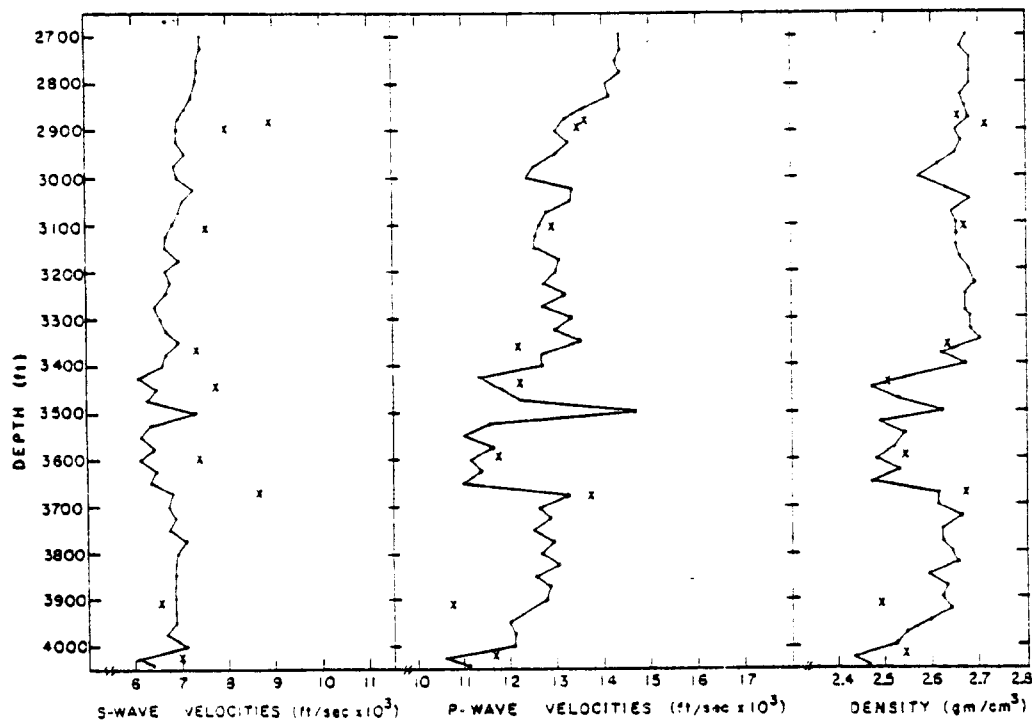


Figure 3. Comparison of laboratory data (Terra Tek, Inc.) with data from the Birdwell (3-D) log on Columbia Gas System Service Corporation well #20403. "X" indicates sample orientation parallel to borehole



TABLE I

Physical Property Measurements  
for Samples from the #20403 Well

IDENT = Sample Identification					% WAT = Moisture (by wet wt.)		
W DEN = Wet Bulk Density (gm/cc)					% POR = Percent Total Porosity		
D DEN = Dry Bulk Density (gm/cc)					% SAT = Percent Saturation		
G DEN = Grain Density (gm/cc)					% A V = Percent Calc. Air Voids		
UBS 2980 A	2.500	2.484	2.600	0.629	4.451	35.349	2.878
2980 B	2.450	2.433	2.630	0.689	7.486	22.553	5.798
2980 C	2.640	2.623	2.680	0.636	2.119	79.210	0.441
MBS 3446 A	2.470	2.459	2.580	0.452	4.697	23.787	3.579
3446 B	2.480	2.468	2.550	0.486	3.210	37.444	2.013
3446 C	2.450	2.437	2.620	0.526	6.981	18.465	5.692
LGS 3760 A	2.650	2.635	2.780	0.561	5.211	18.547	3.724
3760 B	2.630	2.617	2.870	0.495	8.816	14.770	7.514
3760 C	2.640	2.624	2.860	0.600	8.246	19.200	6.663
LBS 3996 A	2.690	2.680	2.750	0.364	2.538	38.573	1.559
3996 B	2.690	2.678	2.870	0.439	6.683	17.677	5.502
3996 C	2.680	2.667	2.710	0.472	1.574	80.353	0.309

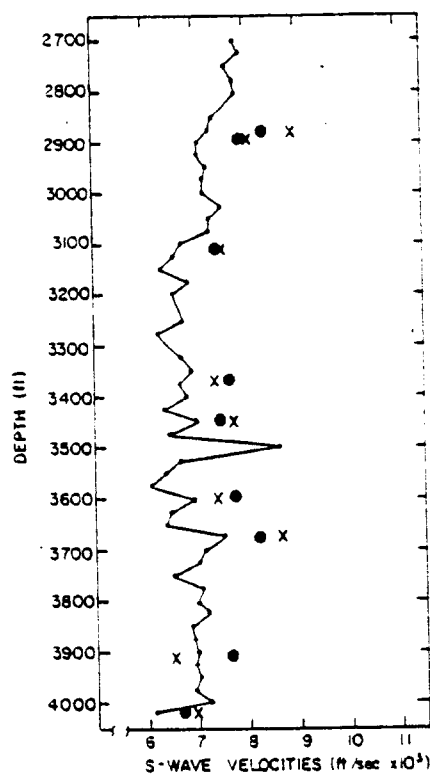


Figure 4. Comparison of log data, laboratory data (x) and calculated s-wave velocities from log data (•), assuming that the log detected the Rayleigh surface wave.

## MECHANICAL PROPERTIES TESTS

Samples from four selected zones of the Devonian shale in well #20403 were used in the mechanical and physical properties tests. The zones are the 2981-2982 ft. interval in the Upper Gray shale, the 3446-3447 ft. interval in the Middle Brown shale, the 3821-2823 ft. interval in the Lower Gray shale and the 3996-3997 ft. interval in the Lower Brown shale. Samples used in the triaxial compression and extension tests were cylinders 0.75 inch in diameter by 1.5 inches in length and were obtained perpendicular to the core axis, i.e., parallel to the bedding plane. Table II lists the results of the measurements and the confining pressures at which the tests were run. The elastic moduli were determined from the initial slopes of the stress-strain curves. The results of these extension tests indicate that the four formations have nearly identical static elastic moduli in extension.

Triaxial compression tests were also carried out to determine the static elastic moduli at simulated downhole conditions. Samples were again taken perpendicular to the core axis and tested under confining pressures adjusted to 1.0 psi/ft. to approximate the overburden stress. A summary of the test data is given in Table III. The unusually high Young's moduli reported for the Lower Brown shale are not believed to be representative of the bulk property of the entire zone. The values are correct for a thin zone from which the samples were extracted but, in this case, do not indicate the average property as do the logs. This was verified by wave velocities measurements on different samples.

As expected, the static moduli are appreciably lower than the moduli derived from sonic velocities.

TABLE II  
Triaxial Extension Test Data for  
Samples from the #20403 Well

Nomenclature	Depth (ft)	Test Density (g/cm <sup>3</sup> )	Confining Pressure (psi)	Young's Modulus (10 <sup>6</sup> psi)	Poisson's Ratio
Upper Brown Shale	2981	2.62	2981	3.43	.27
		2.58	2981	3.00	.13
		2.57	2981	3.00	.11
Middle Brown Shale	3446	2.51	3446	4.00	.20
		2.50	3446	3.33	.20
		2.49	3446	3.43	.14
Lower Gray Shale	3821	2.66	3821	3.00	.20
		2.67	3821	3.15	.23
		2.66	3821	2.86	.17
Lower Brown Shale	3996	2.68	3996	4.00	.20
		2.69	3996	3.19	.20
		2.68	3996	3.61	.20

TABLE III  
Triaxial Compression Test Data for  
Samples from the #20403 Well

Nomenclature	Depth (ft)	Test Density (g/cm <sup>3</sup> )	Confining Pressure (psi)	Young's Modulus (10 <sup>6</sup> psi)	Poisson's Ratio
Upper Brown Shale	2981	2.59	2981	3.75	.11
		2.60	2981	4.80	.28
		2.58	2981	4.29	.23
Middle Brown Shale	3446	2.49	3446	5.25	.17
		2.47	3446	4.50	.20
		2.48	3446	4.62	.23
Lower Gray Shale	3821	2.65	3821	7.20	.35
		2.68	3821	5.00	.23
		2.67	3821	3.60	.14
		2.66	3821	3.30	.27
Lower Brown Shale	3996	2.68	3996	7.50	.41
		2.68	3996	8.00	.41
		2.68	3996	8.50	.23

## FRACTURE TOUGHNESS TESTS

Fracture toughness measurements were carried out on cylindrical samples using the burst technique developed by Clifton, *et al.*, (1976). In this technique a core sample about three inches long is used. A small hole is drilled along the axis of the specimen and two opposite prenotches are placed at the internal walls to specify the fracture initiation points. A bladder is placed in the hole to prevent fluid from entering the sample or the notches and then pressure is applied in the bladder until the sample bursts. Table IV lists the results of these tests.

TABLE IV  
Results of Fracture Toughness Tests

	Sample Depth (ft)	Radii Ratio b/a	Failure Pressure (psi)	$K_{Ic}$ (psi $\sqrt{in}$ )	Remarks
Upper Gray Shale	2711	10.47	2825	1220	Failure initiated along notch then turned along pre-existing fracture.
	2761	10.47	3800	1100	Failure ignored notch and occurred on a pre-existing fracture.
Middle Gray Shale	3365	10.50	2313	737	Fractured as expected.
Lower Gray Shale	3760	10.50	3375	1075	Fractured as expected.

# PERMEABILITY MEASUREMENTS

Permeability measurements were made on three samples from Well #20403 using the pressure transient technique (Appendix C). In this technique a step pressure of a few psi is applied to one side of a sample by a gas (nitrogen) contained in a small reservoir connected with the sample. As the gas penetrates the sample, the pressure in the reservoir declines. The permeability is calculated from the rate of pressure decline. This technique is well suited for permeabilities in the microdarcy range. The results are given in Table V.

TABLE V

Permeability Measurements on Samples  
from the #20403 Well

	Sample Depth (feet)	Confining Pressure (psi)	Pore Pressure (psi)	Permeability (microdarcies)	Comments
Middle Brown Shale	3626	1000	600	8.7	
		4000	600	2.8	
		4000	350	2.7	
Lower Gray Shale	3675	500	350	< 0.1	Permeability was below the resolution of the equipment.
Lower Brown Shale	3965	500	350	15	Sample had visible fractures running the length of the flow path.

Pore pressures of 350 psi were selected to represent the actual downhole pressures. The tests were performed before an accurate measurement of the formation pore pressure was available. Later, however, the actual formation pore pressure was determined by well testing to be 250 psi. The difference between the test pore pressure and the actual formation pressure should have an insignificant effect on the measured permeability.

### FLOW CAPACITY TESTS

A shale sample from the 4,020 foot depth (Lower Brown Shale) in well #20403 was fractured by the "Brazallian" method and fitted back together for the test. Tests were performed on the following arrangements:

1. A fractured sample fitted back together without any proppant. This test was to simulate the part of a hydraulic fracture which was cracked open by the fluid but did not receive any proppant.
2. A one-inch-thick layer of 20/40 sand. This width was selected on the basis of Halliburton's calculation for fracture width for the foam treatment. This test would simulate the part of the fracture that achieved the calculated width and had a sand bed settled in it.
3. A sample propped with 1.44 lb/ft<sup>2</sup> of 20/40 sand to simulate the condition of perfect transport where the suspended sand remained in place while the fracture closed.
4. A sample propped with 0.56 lb/ft<sup>2</sup> of 100 mesh sand to simulate a spearhead of fine sand in the fracture which would behave mostly like a fluid-loss material.

These tests were made at several confining pressures in order to bracket the actual *in situ* closure pressures. The sand, sent to Terra Tek by Halliburton, was a sample of the material used in the hydraulic fracture treatment in Well #20403. A description of the experimental procedure is given in Appendix D. The results are summarized as follows:

1. Unpropped Fracture. Flow capacities were dependent on both the closure stress on the rock faces and the pressure of the gas used for the measurements. For downhole closure stresses between about 1,000 and 4,000 psi, and dry nitrogen pressures between about zero and 100 psi the flow capacities ranged from about 1 md-cm to about 300 md-cm. Visual examination of the faces of the samples after the test found only small contact marks. There was no visible plastic flow or imbedment. An approximate calculation (Appendix D) of the effective open fracture width is 0.001 to .03 cm for downhole closure stress conditions.
2. Fracture Propped with 0.56 lb/ft<sup>2</sup> of 100-mesh sand. Again, the flow capacities were dependent on closure stress and gas pressure. Measurements at simulated downhole closure stress, but low gas pressure yielded flow capacity measurements of about 1100 md-cm.
3. One-inch-thick Layer of 20/40 Sand. The sand did not crush under the simulated downhole stresses so the measurements were essentially those for compacted sand. The permeability to dry nitrogen measured at pressures near atmospheric pressure was in the range to 6,000 to 10,000 md-cm, and there was only a small decrease in permeability with increasing confining stress.
4. Fracture Propped with 1.44 lb/ft<sup>2</sup> of 20/40 Sand. Flow capacities were dependent on both the closure stress



and the gas pressure. Plots of flow capacity versus reciprocal pressure did not always show a linear relationship as expected from the Klinkenberg theory. For downhole conditions of pressure and closure stress the flow capacity is approximately 8,000 md-cm.

Figure 5 is a plot showing the results for the unpropped fracture and for the fracture propped with 0.56 lb/ft<sup>2</sup> of 100 mesh sand and 1.44 lb/ft<sup>2</sup> of 20/40 mesh sand. Because of the pressure and flow rate dependency, the plots are indicated as bands with trends rather than definite lines.

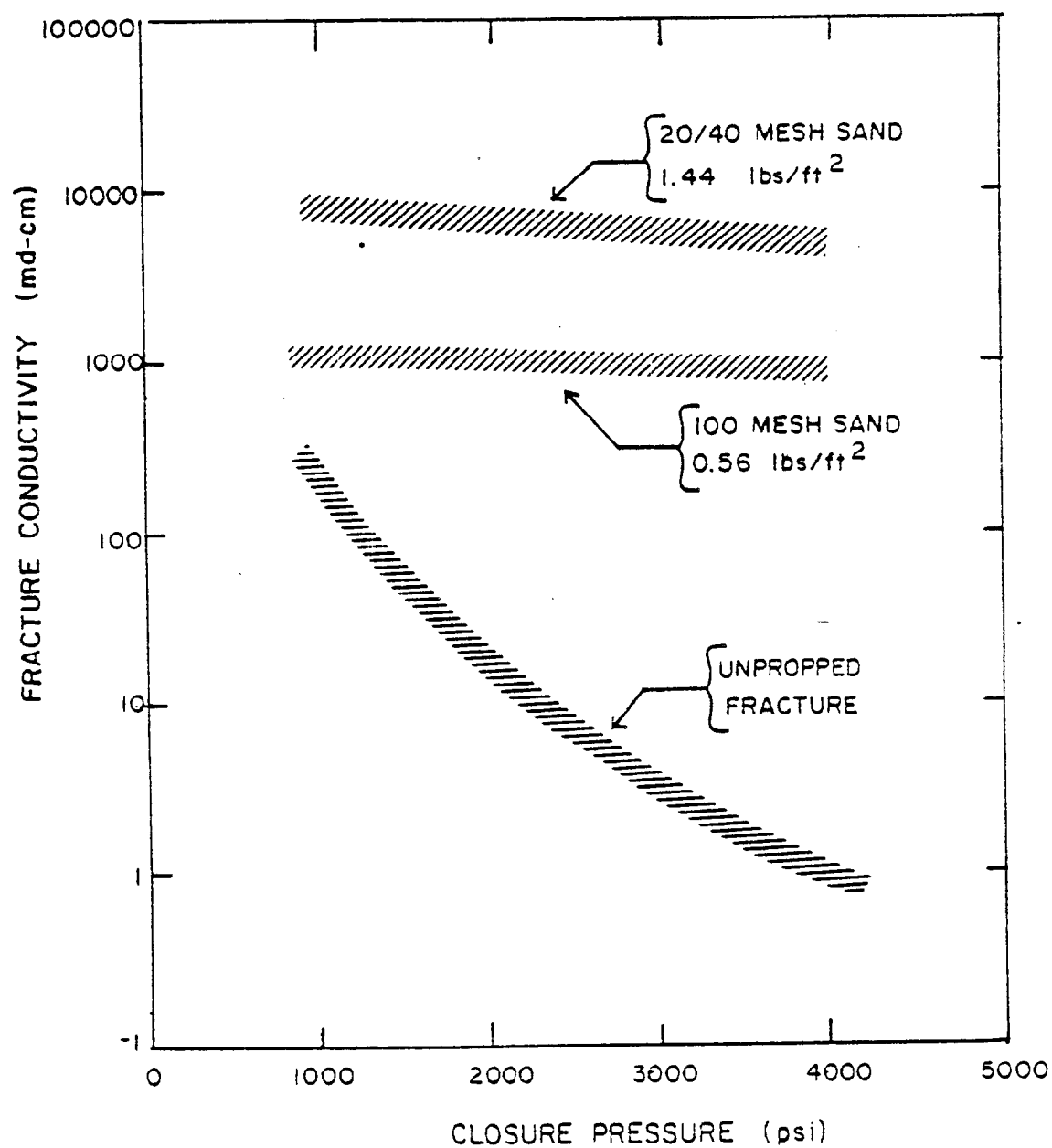


Figure 5. Fracture conductivity for unpropped and propped fracture in Lower Brown Shale.

## DATA INTERPRETATION AND CONCLUDING REMARKS

### BACKGROUND

The Devonian shale in the area of the three-well test program consists of several layers of brown and gray shales, as shown in Figure 1. Natural gas is found in all of the layers with the Middle and Lower Brown Shale layers having the best potentials. Since the bedding is nearly horizontal and there are no major tectonic displacements in the area of the test program, the shale layers are expected to be similar in the 20401, 20402 and 20403 wells. Data from all three wells are thus combined to form a composite picture of the area.

In hydraulic fracturing of reservoir rock formations, the rock is subject to:

- 1) The fracturing fluid pressure which tends to open the crack, and
- 2) The far field *in situ* stresses acting to close the crack.

During a hydraulic fracture treatment, the forces from the fracturing fluid need to overcome both the *in situ* stress and the fracture toughness of the rocks, a material property similar to the tensile strength that represents the resistance of rock to fracturing. In a layered subsurface formation such as the Devonian shale, the separate layers may have different fracture toughness, and/or elastic moduli along with different naturally occurring stresses in the ground. Fracture containment analysis is accomplished by evaluating these factors as they apply to the particular zone of the gas reservoir being stimulated. To determine whether a layer is a barrier or not, it is thus necessary to evaluate its properties in relation to the adjacent layers. A particular layer may be a barrier in some instances and not a barrier in other instances, depending on the relative relationships of the stresses and mechanical properties.

## CONTAINMENT CRITERIA

When a fracture is initiated in rock, the fracture will extend when the stress-intensity factor at its tip reaches a critical value. Whether a fracture moves up, down or out depends on the relative values of the critical stress-intensity factors for the various materials as well as along the fracture perimeter as generated by the loading conditions and fracture geometry. Hence, a fracture analysis requires the determination of the stress-intensity values around the crack periphery and then their comparison to the critical values for the different formations [Simonson, *et al.*, 1975]. The following data is used in such analysis:

- 1) The elastic moduli in the pay zone and bounding layers
- 2) The *in situ* stress field and, in particular, the minimum principal stresses in the pay zone and bounding layers above and below the pay zone
- 3) The fracture toughness or critical stress-intensity factors for the pay zone and bounding layers
- 4) The physical properties of the fracturing fluid and the pumping schedule.

Knowledge of the elastic moduli and *in situ* stresses for the pay zone and both of the bounding layers is needed to determine whether or not the bounding layers are barriers to crack extension outside the pay zone.

Hydraulic fracture analysis is inherently a three-dimensional problem; the mathematical solutions of which are extremely complicated. The present work will be limited to treating two-dimensional cracks. Such a simplified analysis provides considerable insight into understanding the parameters and conditions which influence hydraulic fracture propagation [Simonson, *et al.*, 1976].

The basic concept of containment analysis is illustrated in Figure 6. The fluid is moving such that the pressure near the wellbore is higher than the pressure near the tip. Estimates of the stress-intensity factor at the edges and tip of the propagating fracture can be made if the pressure profile, *in situ* stresses and elastic moduli along the crack shape are known. These estimates are then compared to the critical stress intensities. The fracture will propagate when the stress-intensity factors reach the critical values. In the special case where the stress-intensity factor would be equal to the critical value at all points of the perimeter around the crack, the fracture would propagate in all directions at the same time. On the other hand, if the stress-intensity factor reaches a critical value only at the tip and not at the upper and lower edges, the fracture is then confined to the pay zone.

For this discussion we will consider an infinitely long, narrow crack in layered media. This model is representative of a hydraulic

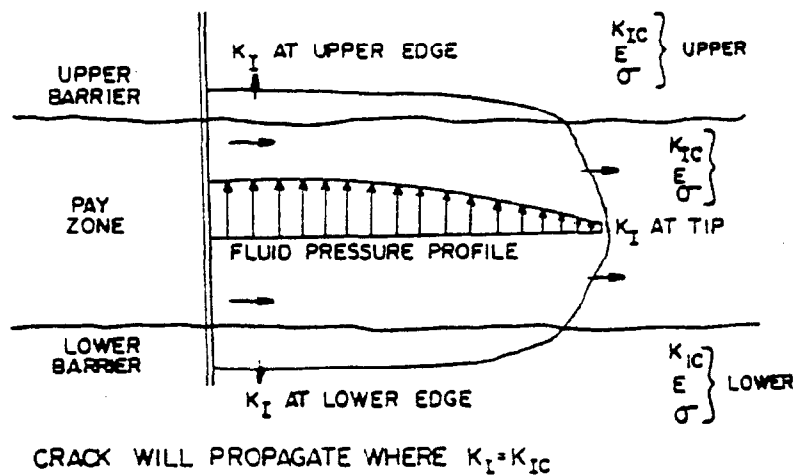


Figure 6. Schematic diagram to illustrate fracture containment analysis

fracture with a short height and a long length. The plane-strain mathematical model gives the following equation for the stress-intensity factor at the upper crack edge

$$K_I = \frac{1}{\sqrt{\pi l}} \int_{-l}^{+l} (P_y - S_y) \sqrt{\frac{l+y}{l-y}} dy$$

where

$P_y$  = Pressure profile in crack

$S_y$  = Far-field horizontal stress profile

$y$  = Position along crack

$l$  = Height from the crack center to the tip

$S_y$  represents the minimum principal *in situ* stress since fracture will be normal to this stress. Note in this equation that the stress-intensity factor is dependent upon the profile of the stress difference between the pressure inside the fracture and native *in situ* stresses across the fracture. Different combinations of fluid pressures and *in situ* stresses will give different values of  $K_I$  and, in turn, different results for a prediction of the final fracture geometry.

#### (i) Influence of Elastic Moduli

The above equation for  $K_I$  is for the hypothetical case where the elastic moduli of the pay zone and the bounding layer are equal. If the elastic moduli of the several layers were different, then there is an additional influence on the stress-intensity factor as a result of the difference in the moduli. Figure 7 illustrates the qualitative behavior

of the stress-intensity factor for a crack that moves perpendicular to an interface between two materials with different moduli when plane-strain conditions prevail. If the elastic modulus in the pay zone is higher than that of the bounding layer, then as a fracture propagates toward the boundary, the stress-intensity factor increases such that it will reach the critical value and the fracture will "snap" into the bounding layer. On the other hand, if the modulus in the pay zone is lower than the moduli in the bounding layers, then a fracture which is propagating from the pay zone toward the bounding layers will find the stress-intensity at its tip nearest the interface decreasing as the boundaries are approached such that the critical value is not reached and the fracture is contained within the pay zone.

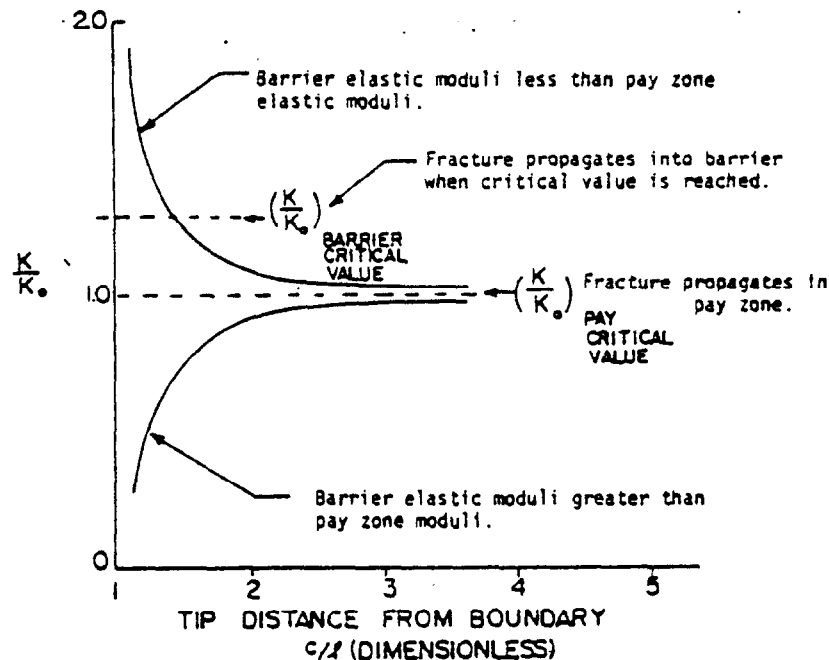


Figure 7. Qualitative behavior of the stress-intensity factor as a crack of length  $2a$  approaches the boundary of an adjacent layer with different elastic moduli ( $c$  is the distance from fracture center to boundary)

(ii) Influence of *In Situ* Stress

Difference in *in situ* stresses between the pay zone and the barrier layers have a distinct influence on fracture propagation. Consider the hypothetical case, shown in Figure 8 in which a hydraulic fracture has extended by some mechanism or other into adjacent layers with similar

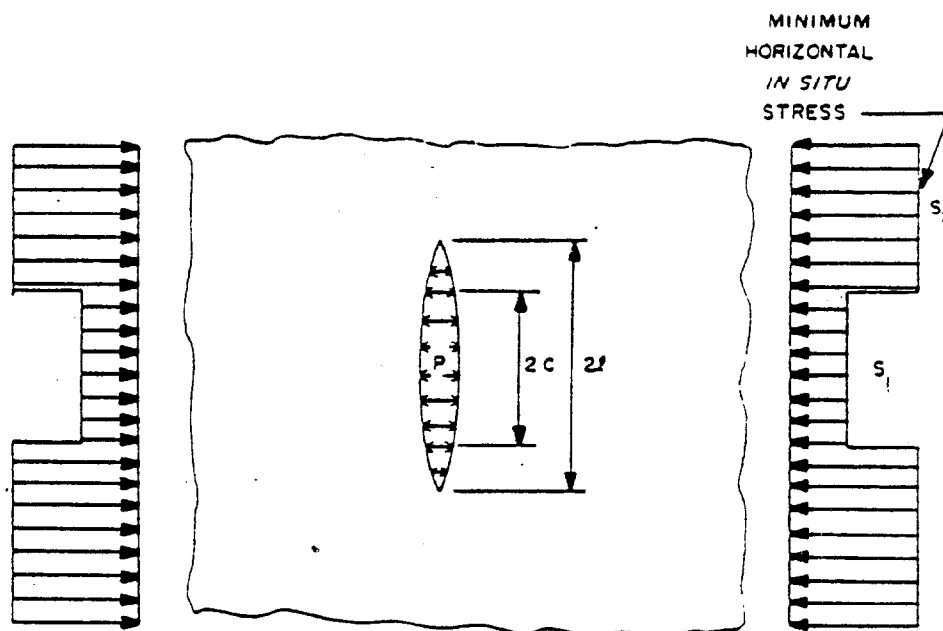


Figure 8. Vertical hydraulic fracture loaded under uniform pressure ( $P$ ) with differing minimum horizontal stress in pay zone and barriers

elastic moduli but different *in situ* stresses. Figure 9 shows a plot of the distance the crack will advance into the region of high stress (the barrier layers) in terms of the pressure,  $P$ , within the fracture and  $P_0$ , the fracture fluid pressure required for the fracture to reach the interface. The curves in this figure are for a crack height of 200 feet, a fracture toughness of  $1,000 \text{ psi}\sqrt{\text{in}}$  and for parametric values of the *in situ* stress difference  $S_2 - S_1$ . For a stress difference of 1,000 psi, for example, an over pressure of 500 psi would be expected if the fracture were



to propagate a distance of 100 feet into the region of higher *in situ* stress. Finally, if the *in situ* stress in the barrier layers ( $S_2$ ) were less than the *in situ* stress in the pay zone ( $S_1$ ), a situation would exist where it requires less pressure to propagate the fracture in the barrier than in the pay zone.

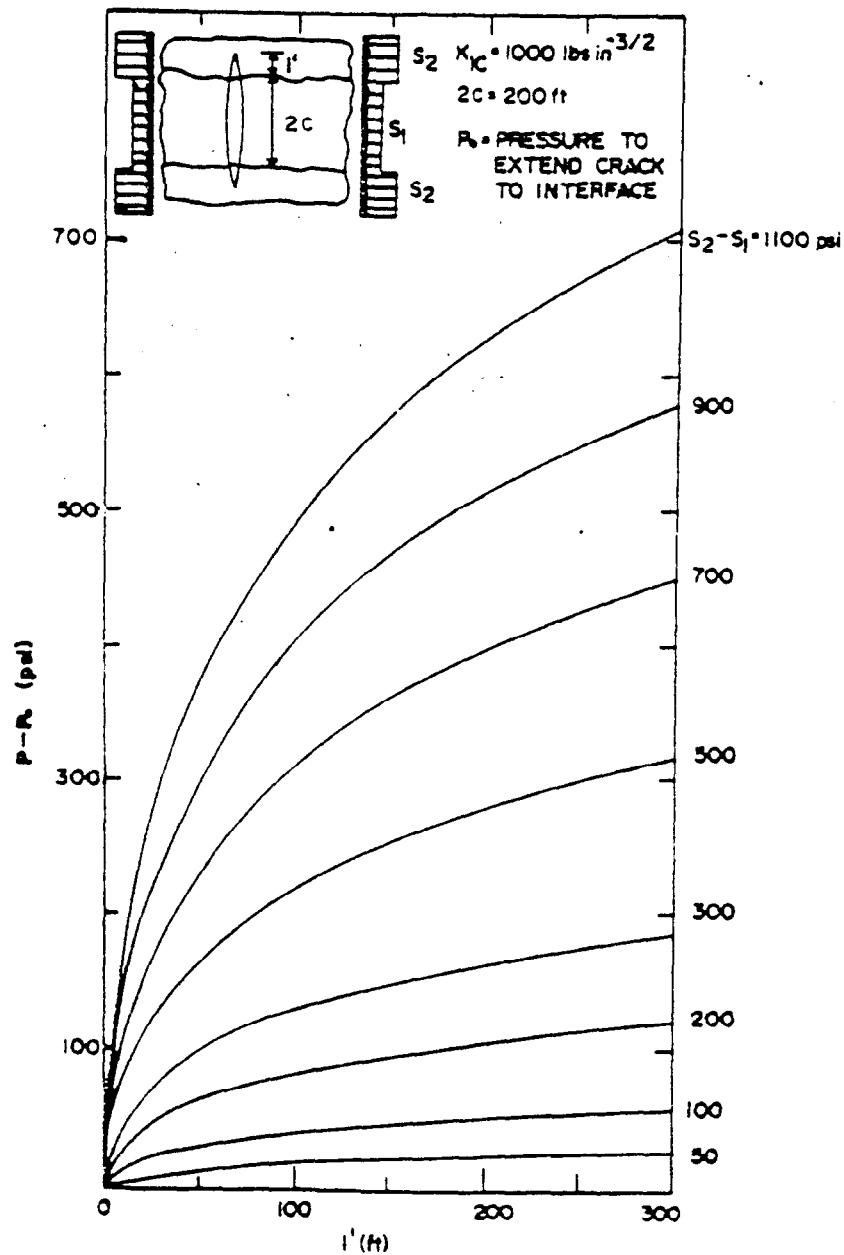


Figure 9. Estimate of fracture migration into the barrier layers [from Simonson, *et al.*, 1976]

## SUMMARY OF ROCK MECHANICS DATA

Young's moduli have been calculated using the measured bulk densities. P-wave velocities and the revised S-wave velocities as an incomplete set of triaxial data is available. For the purpose of these calculations, the material was assumed isotropic and homogeneous. Figure 10 illustrates the general trend of the data. Also shown are static moduli from triaxial compression of the Upper Brown and Lower Gray Shales. As expected, the static moduli are lower than the sonic moduli; for the sample tested, the difference is on the order of 20 to 30 percent. Figure 10 summarizes the results of critical stress-intensity tests for the Gray Shales.

*In situ* stresses, especially the difference in *in situ* stresses between the pay and barrier formations, are the most critical parameters to MHF containment. Unfortunately, this is the parameter for which the least amount of data is available. A single mini-hydraulic fracturing test was performed by Terra Tek at the 2745 foot level (Upper Gray Shale) in well #20402 [Abou-Sayed, *et al.*, 1977]. These results were as follows:

Minimum Horizontal Stress	2360 psi, N 35° W to N 45° W
Maximum Horizontal Stress	4390 psi, N 45° E to N 55° E
Overburden Stress	3210 psi
Pore Pressure	~ 250 psi

The measured principal stress directions agree well with the prevailing geological structure in the region and with the principal stress directions reported by Overbey (1976) from a series of measurements in West Virginia. A basement structure map of the region in which the reported tests were conducted is shown in Figure 11. The map, adapted from Harris (USGS Map I-919D, 1975) by Overbey (1976), shows a projection of the Rome Trough

the projection of the Rome Trough are N 45° E to N 50° E, i.e., parallel to the strike of the basement faults and in agreement with the directions reported here.

An estimate of the minimum horizontal stresses in the other shale layers was made on the basis of elastic theory. Data from the Upper Gray Shale was analyzed for determination of applied boundary displacements in both horizontal directions as might be imposed through normal faulting on the boundaries of the Rome Trough. These same displacements were applied to the other shale layers and the resulting minimum *in situ* stress in these layers was estimated using moduli derived from corrected logs. Figure 10c shows a plot of the estimated minimum horizontal principal stresses in the different formations. These estimates correlate with two independent observations. Swolfs, *et al.*, (1977), based on strain relaxation data, estimated the minimum *in situ* stress gradients from the surface of 0.52 psi per foot for the Middle Brown Shale and 0.76 psi per foot for the overlying Middle Gray Shale. These stress gradients coincide with the data shown in Figure 10c at the Middle Gray/Middle Brown Shales interface. Field engineers contend that either the Lower Gray or the Lower Brown Shale will break down under a hydrostatic head of water. The estimated minimum *in situ* stress shown in Figure 10c for the Lower Brown Shale would also predict this behavior.

## DATA INTERPRETATION

Although the tests were minimal, sufficient data was gleaned from available information to perform a containment analysis. Both Figure 10a and 10c indicate that the gray shales and Onondaga Limestone will act as barriers to the fracture growth in the brown shales. The gray shales are not strong barriers (moduli contrast is less than 30 percent and the largest *in situ* stress difference is less than 800 psi), however, and can also be broken down if fluid pressures are too high. The brown shales will not form barriers to fractures in the gray shales. This interpretation correlates with field results reported by McKetta (1977). A fracture initiated in the Lower Gray Shale in well #20403 propagated upward into the Middle Brown Shale. Communication was noted on perforating and breaking down the Middle Brown Shale.

A significant factor in successful completion of MHF treatment in reservoirs with marginal barrier formations is the control of fracturing fluid flow rate and the maximum allowable bottomhole treatment pressure (BHTP). There are other factors, besides containment of the induced fracture, that might impose certain bounds on both flow rate and BHTP. These are proppant transport, created crack width, strength of casing, location of perforation and the like. However, the calculated value of BHTP that would prevent the fracture penetration into barrier zones would prove instrumental in achieving a contained fracture geometry. An estimate of the maximum BHTP for a given treatment can be obtained following the arguments presented earlier. For a pay formation height of  $2c$  and a pumping pressure of  $P$ , the fracture will extend a length  $z'$  into the barrier layer such that [Simonson, *et al.*, 1976]:

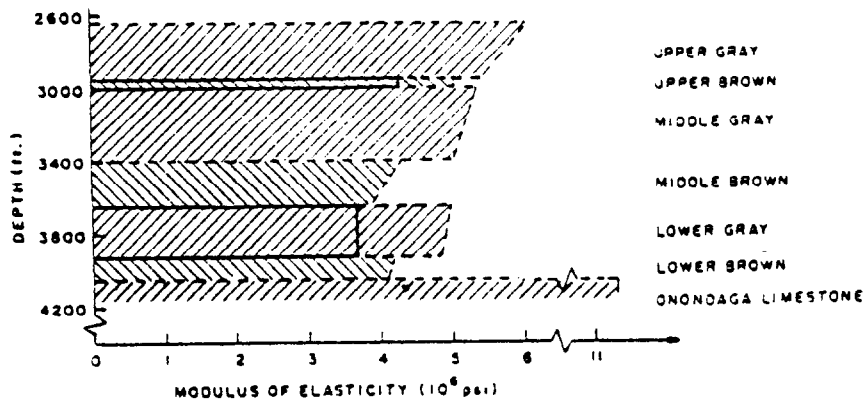


Figure 10a. Log-derived (dotted lines) and laboratory-measured (solid lines) modulus of elasticity

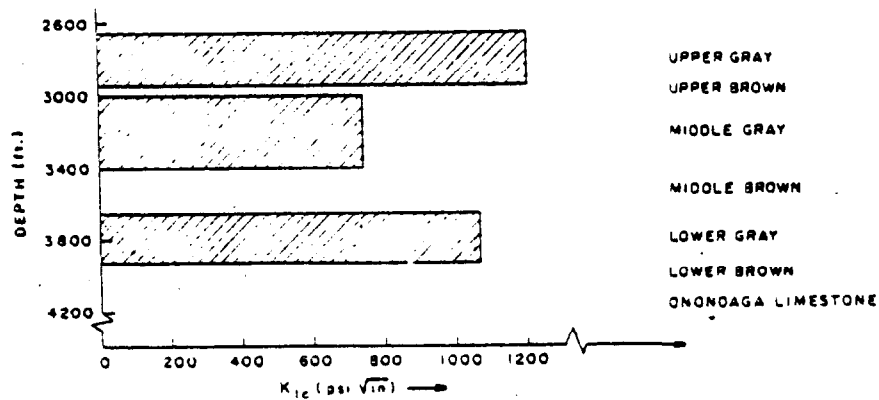


Figure 10b. Critical stress-intensity factor for the barrier shales only

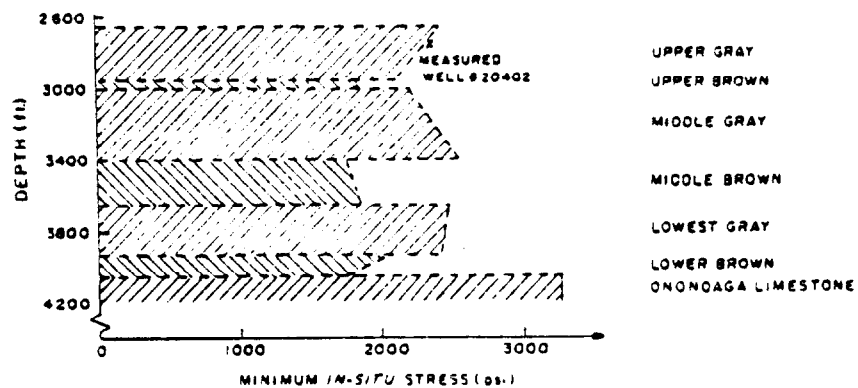


Figure 10c. Derived minimum *in situ* stresses correlated to measured stresses (x) at 2745 ft. in Well #20402

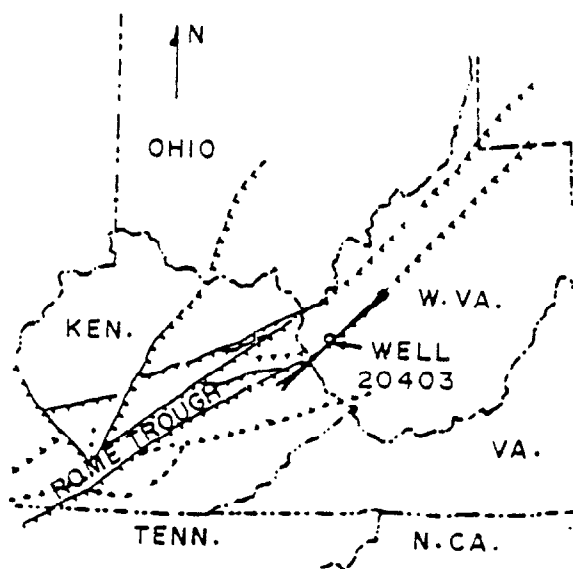


Figure 11. Basement structure of Kentucky-West Virginia adapted from Overbey (1976) with the direction of maximum horizontal stress at well #20403

through the northwestern edge of West Virginia. Schumaker (1976) describes the Rome Trough as a graben bounded by high-angle, normal faults. This structure lies within an area which is a junction of three distinct geological provinces [Werner, 1976]:

- 1) Central Appalachian Fold Belt,
- 2) Southern Appalachian Thrust-Fold Belt, and
- 3) Appalachian Plateau with Basement Faults.

Although it is not known whether or not basement faults penetrate into the Devonian shales in this area, Overbey's measurements (1976) suggest a correlation of principal stress directions with the basement structure. At locations which fall outside the projection of the Rome Trough, the measured directions of  $\sigma_{HMAX}$  are oriented generally east-west and are normal to the predominantly north-south strike of the thrust faults and folds of the Appalachian Mountains. Measured directions of  $\sigma_{HMAX}$  which fall within

$$P - S_1 = \frac{1}{\sqrt{\pi(c+l')}} \left[ K_{Ic} - \frac{2(S_2 - S_1)(c+l') \cos^{-1} \left( \frac{c}{c+l'} \right)}{\sqrt{\pi(c+l')}} \right]$$

where  $K_{Ic}$  is the critical stress-intensity factor for the barrier layer and  $S_1$  and  $S_2$  are the minimum *in situ* stresses for the pay and barrier zones, respectively. Penetration of a fracture which initiated in the Middle Brown Shale into the Middle Gray Shale is shown in Figure 12. For excess pumping pressure in the fracture ( $P - S_1$ ) less than 400 psi, the penetration is kept within 50 feet. To maintain fracture intrusion into the Middle Gray Shale within this limit, the bottomhole treatment pressure during hydraulic fracturing of the Middle Brown Shale layer should not exceed the minimum horizontal stress in that zone by more than 400 psi, plus any expected drop in pressure across the perforation. Corresponding considerations for the Lower Brown and Lower Gray Shales (the Onondaga Limestone forms a strong barrier due to its large elastic moduli) leads to a treating pressure within the formation of 250 psi in excess of the minimum horizontal stress for the Lower Brown Shale. For the Upper Brown Shale, the Upper Gray and Middle Gray Shales will form weak barriers. Following similar considerations as outlined above, this leads to an in formation treating pressure of 200 psi in excess of the minimum horizontal stress for fracturing the Upper Brown Shale.

In treating the Middle Brown Shale first, the fracture will propagate into both the Middle Gray and Lower Gray Shales. There is, therefore, the possibility that later fractures initiated in the Upper Brown and in the Lower Brown Shales will interact with the earlier fracture created in the Middle Brown Shale. In a multi-stage MHF it is important that the better reservoir formation be fractured first.

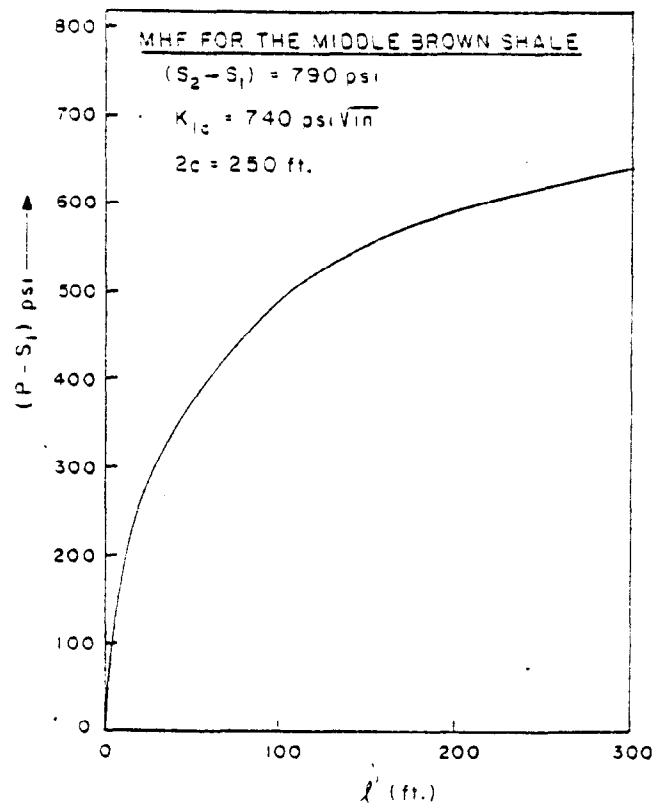


Figure 12. Fracture penetration into the Middle Gray Shale for fracture in the Middle Brown Shale



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APPENDIX A

COMPARISON OF LOG AND LABORATORY MEASURED  
P-WAVE AND S-WAVE VELOCITIES

Terra Tek, Inc.  
Report TR 77-19

Accepted for presentation at the  
Eighteenth Annual Logging Symposium

## COMPARISON OF LOG AND LABORATORY MEASURED P-WAVE AND S-WAVE VELOCITIES

by

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### ABSTRACT

Longitudinal (p-wave) velocity, shear (s-wave) velocity, and density measurements were made in the laboratory under simulated *in situ* conditions, on Devonian shale core samples. These values and calculated elastic moduli are compared with data obtained from wire-line logs. The densities and p-wave velocities are in good agreement; however, the laboratory s-wave velocities are consistently higher (10 to 15 percent) than the log measurements. These discrepancies are greatly magnified in the derived elastic moduli. Hypothesizing that the log-detected s-wave is in reality the Rayleigh wave, the calculated s-wave velocities agree as well as the p-wave velocities with the corresponding lab-measured values.

### INTRODUCTION

Many engineering applications now require knowledge of *in situ* formation elastic properties. Well completion, production scheduling in fields with weak formations, fluid injections to increase recovery, and massive hydraulic fractures are a few of the applications. Since the elastic moduli can be calculated from material density, p-wave velocity ( $V_p$ ), and s-wave velocity ( $V_s$ ), this work was directed toward the comparison of these data obtained from wire-line logs and simulated *in situ* laboratory measurements. The measurements were made on Devonian shale cores obtained from Columbia Gas System Service Corporation well #20403 located in Lincoln County, West Virginia. This well was logged with both the Schlumberger experimental tool and the Birdwell 3-D tool.

In general, the data from the logs agreed with each other and both agreed with the laboratory density and p-wave velocity measurements. However, the s-wave velocities obtained on the laboratory samples were consistently higher (10 to 15 percent) than those obtained from the logs. Errors in the derived elastic moduli are greatly magnified. For example, a 10 percent difference in the s-wave velocity would produce over a 30 percent change in the calculated Poisson's ratio.

The reason for the difference in the s-wave velocities is not known. However, the consistent offset in the s-wave data along with the good agreement of the p-wave velocities introduces a question as to what was actually being measured by the logs. It is hypothesized that the log measured s-wave is in reality the Rayleigh wave.

## LABORATORY TECHNIQUES

The laboratory test samples were prepared by coring parallel to the original core. They were 1 inch in diameter and typically 1.5 inches long, the ends were ground flat and parallel to within .001 inch. A urethane jacket sealed to the end cap was used to prevent confining fluid contact with the rock. The sample was mounted to the removable base plug and the entire assembly inserted into the pressure vessel, which was subsequently filled with fluid and pressurized. Feedthroughs were provided to enable electrical contact with the ultrasonic transducers. The test configuration is shown in Figure 1.

A through-transmission technique similar to that introduced by Mattaboni and Schreiber (1967) was used in the laboratory to obtain the velocities. A single transmitting transducer (1 MHz) was used to produce both a p-wave and an s-wave. Separate receivers detect the arrival of each signal. The accuracy of this technique is better than 0.5 percent.

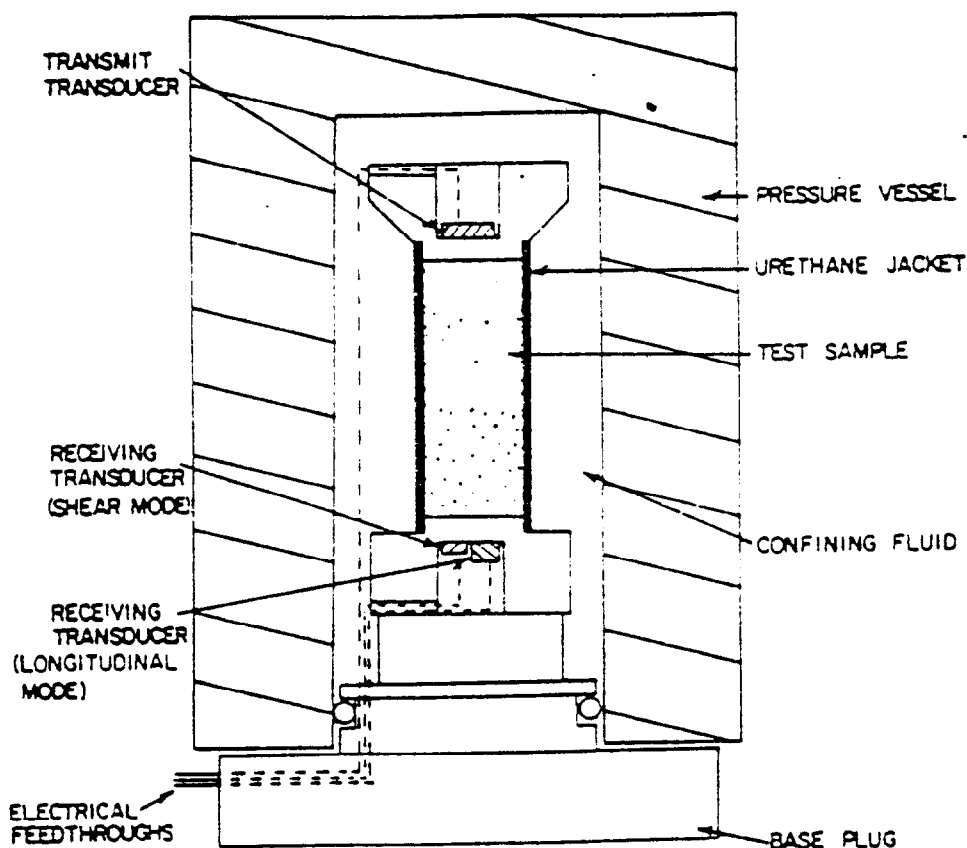


Figure 1. Test configuration for laboratory ultrasonic measurements.

Certain precautions were taken to insure the validity of comparing laboratory and log data. For example, all velocity measurements were made in the same plane and with the same s-wave polarization as the logs, this avoided complications resulting from material anisotropy. Also all measurements were performed at simulated *in situ* stresses. Based on the measured *in situ* stresses, [Brechtel, et al. (1976)] hydrostatic conditions are a close approximation of the measured stress, with a gradient of 1 psi/foot.

The effect of confining pressure on the measured velocities was very small, as indicated by Figure 2. The sample from 3,365 feet indicates the greatest change but even for this sample the p-wave velocity varied less than 6.5 percent for a 4000 psi change in confining pressure. The corresponding change in s-wave velocity is slightly more than 3.5 percent for the same change in pressure.

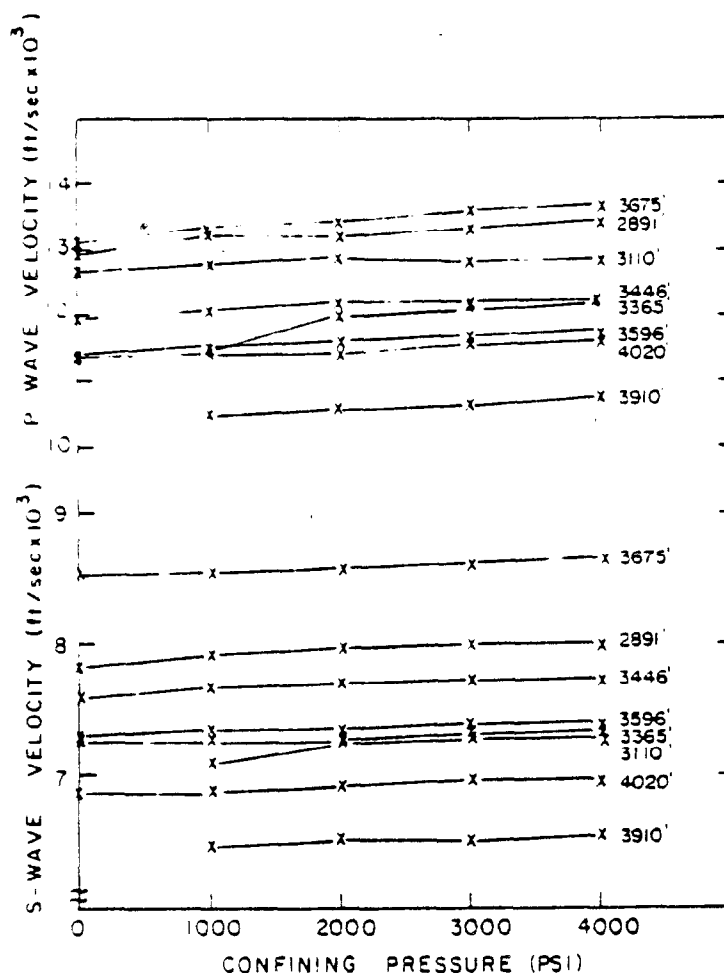


Figure 2. P-wave and s-wave velocities as functions of hydrostatic confining pressure on Devonian shale samples from various depths.

Preliminary measurements showed very little velocity change with temperature over the range encountered in this well ( $\sim 100^{\circ}\text{F}$ ). The change in the p-wave velocity was less than one percent and the s-wave velocity was within the accuracy of the measuring system. This is consistent with reported velocity versus temperature data on geological materials, c.f. Timur (1976). Therefore, all laboratory velocity data reported here were taken at room temperature.

Densities were obtained from the measured dimensions and weight of the prepared test samples. The accuracy of these measurements were within one percent. The measured change in the density with applied confining pressure up to 4000 psi was no more than 0.2 percent. Because of this small change, the data was not corrected for change in density with pressure.

### COMPARISON OF DATA

Figures 3 and 4 shows the comparison of laboratory data with that obtained from the logs. The p-wave velocities and densities are, in general, in good agreement, with the exception of the sample from 3910 feet. The laboratory s-wave velocities are consistently 10 to 15 percent higher than the log values at identical depths. Young's modulus, Poisson's ratio, and the bulk modulus have been calculated, using the measured bulk densities, p-wave velocities, and the s-wave velocities. For the purpose of these calculations, the material was assumed isotropic and homogeneous. The equations that were used are presented in Appendix A1. The comparison of these data are shown in Tables I and II. As anticipated the discrepancy in s-wave velocities are greatly magnified in the elastic moduli.

With the geometry of a sonic log in a borehole, it is physically possible to generate, by mode conversion at the fluid-rock interface, a Rayleigh wave (surface wave) in addition to the p- and s-wave. This is shown in Figure 5. One possible explanation for the discrepancy between the laboratory and log s-wave velocities is that the logs were detecting the arrival of the Rayleigh wave ( $V_r$ ) and not the s-wave. If this assumption is made then it is possible to calculate the s-wave velocities using the established relationships found in the open literature, c.f. Fung, Chap. 7, p. 180 (1965). Because of the difficulties in solving the equation directly for the Rayleigh wave velocity, a plot relating the velocity ratios was made and is presented in Figure 6.

The revised s-wave velocities, calculated from the Schlumberger log data, are shown in Table III. The comparison of the revised data, original log data, and the laboratory data is shown in Figure 7. With the exception of the 3910 foot samples, the revised s-wave velocity values agree very well (within 0.5 percent on the average) with those obtained from laboratory measurements.

### CONCLUSIONS

The close agreement obtained on the measured p-wave velocities and densities is a good indication that the laboratory samples were representative of the material encountered by the logs. The data strongly indicates that the logs measured the arrival of the Rayleigh wave and not the s-wave. It is obvious that more work is needed in this area in order to properly assess the

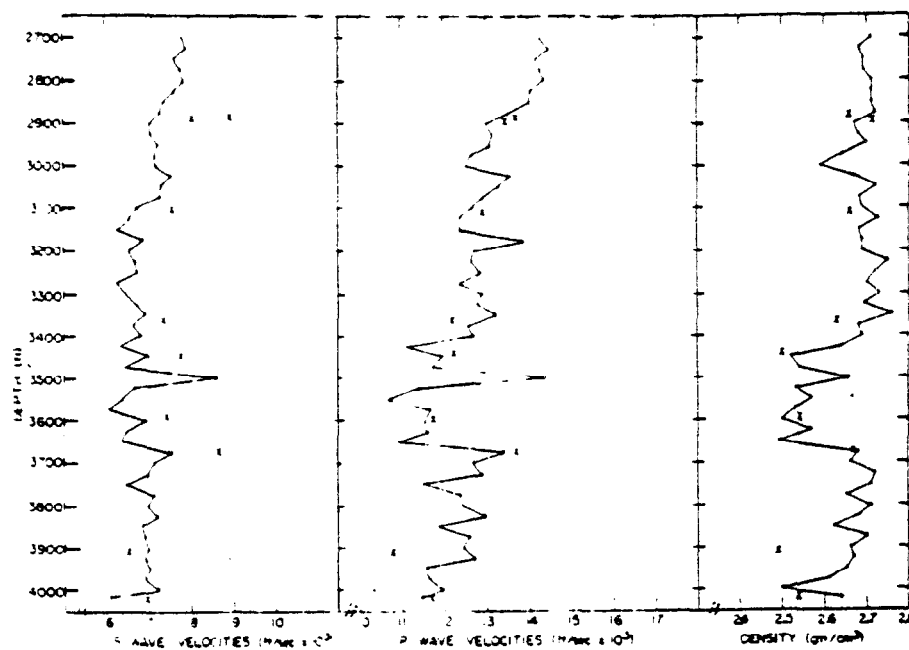


Figure 3. Comparison of laboratory data (X) with data from a Schlumberger experimental s-wave logging tool.

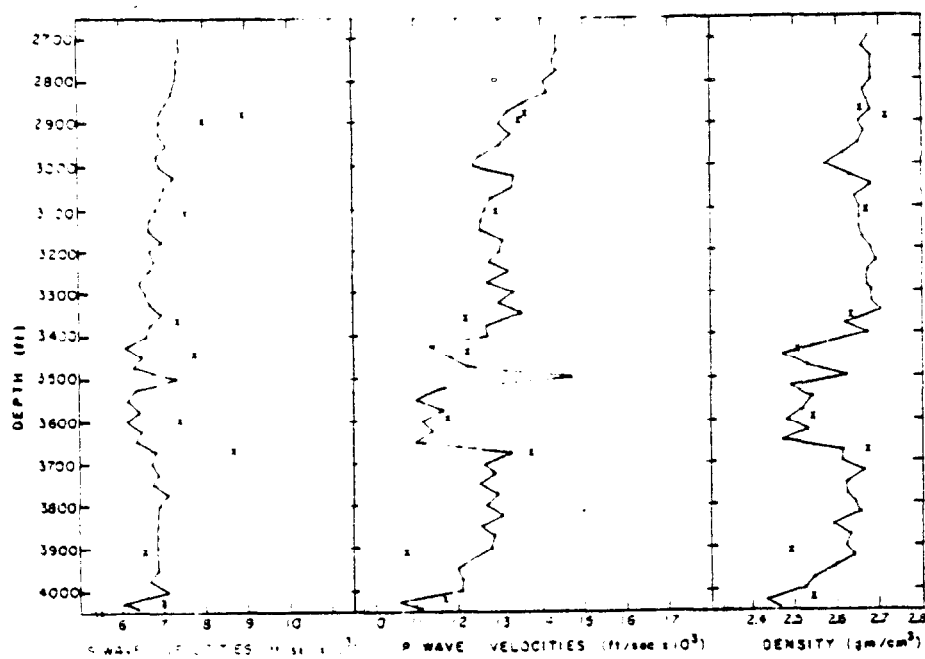


Figure 4. Comparison of laboratory data (X) with data from the Birdwell (3-D) log.



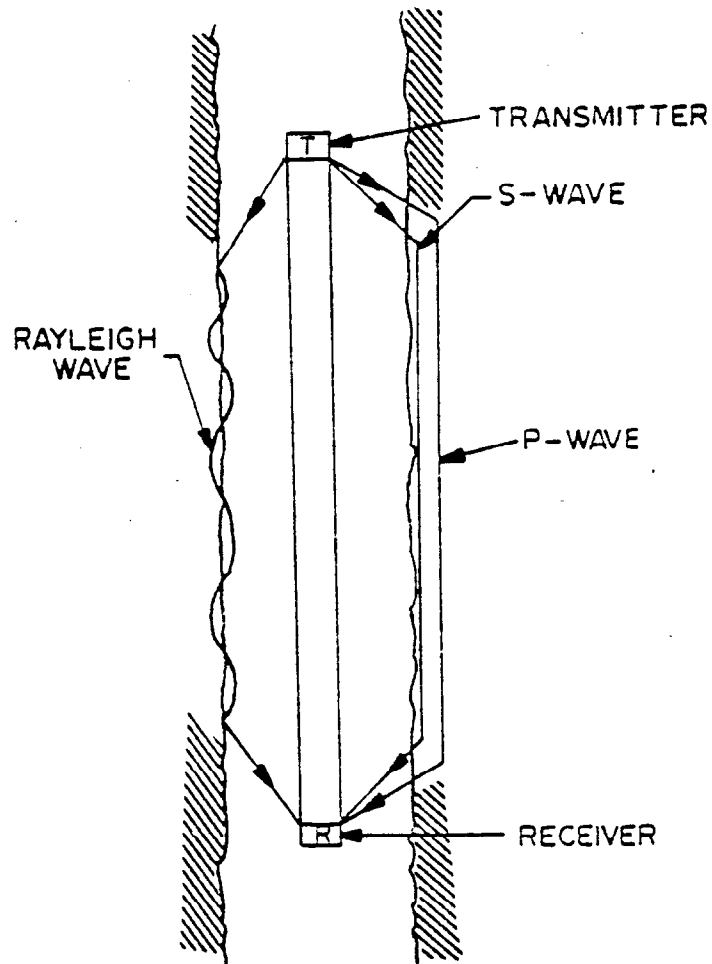


Figure 5. Ray paths for elastic wave produced by a sonic tool in a fluid-filled borehole.

TABLE I

Laboratory Data (Terra Tek, Inc.) and Log Data (Birdwell)  
on Columbia Gas System Service Corp., Well #20403 (Devonian Shale)

DEPTH (FT.)	VELOCITIES				MODULI						POISSON'S RATIO		DENSITY $\rho$ GM/CM <sup>3</sup>	
	P-WAVE (FT/SEC)		S-WAVE (FT/SEC)		YOUNG'S E (PSI $\times 10^6$ )		BULK K (PSI $\times 10^6$ )		SHEAR G (PSI $\times 10^6$ )					
	LOG	LAB	LOG	LAB	LOG	LAB	LOG	LAB	LOG	LAB	LOG	LAB	LOG	LAB
2880	13420	13640	7180	8950	4.31	6.43	4.00	2.34	1.35	2.36	.300	.122	2.67	2.66
2891	13460	13450	7270	8030	4.30	5.75	3.96	3.46	1.89	2.35	.294	.223	2.66	2.71
3110	12490	12900	5830	7560	4.27	5.06	3.34	3.23	1.66	2.04	.287	.239	2.64	2.66
3365	11650	12170	6670	7340	3.89	4.63	2.66	2.70	1.55	1.91	.257	.214	2.58	2.63
3446	12040	12230	6330	7740	3.56	4.69	3.10	2.35	1.36	2.01	.309	.167	2.52	2.50
3596	11200	11740	6080	7400	3.19	4.39	2.55	2.22	1.24	1.87	.291	.170	2.48	2.54
3675	13250	13710	6800	8660	4.31	6.30	4.01	3.15	1.63	2.70	.321	.168	2.61	2.67
3910	12280	10780	6760	6540	4.16	3.47	3.19	1.99	1.62	1.44	.283	.209	2.63	2.49
4020	11880	11670	6600	6970	3.35	4.07	2.87	2.45	1.51	1.66	.276	.223	2.57	2.54

TABLE II

Laboratory Data (Terra Tek, Inc.) and Log Data (Schlumberger)  
on Columbia Gas System Service Corp., Well #20403 (Devonian Shale)

DEPTH (FT.)	VELOCITIES				MODULI						POISSON'S RATIO		DENSITY $\rho$ GM/CM <sup>3</sup>	
	P-WAVE (FT/SEC)		S-WAVE (FT/SEC)		YOUNG'S E (PSI $\times 10^6$ )		BULK K (PSI $\times 10^6$ )		SHEAR G (PSI $\times 10^6$ )					
	LOG	LAB	LOG	LAB	LOG	LAB	LOG	LAB	LOG	LAB	LOG	LAB	LOG	LAB
2880	13560	13640	7630	8950	5.39	6.43	3.88	2.34	2.13	2.36	.268	.122	2.71	2.66
2891	13330	13450	7260	8030	4.91	5.75	3.88	3.46	1.90	2.35	.289	.223	2.68	2.71
3110	12580	12900	6800	7560	4.32	5.06	3.49	3.23	1.67	2.04	.293	.239	2.68	2.66
3365	11430	12170	6870	7340	4.07	4.63	2.40	2.70	1.67	1.91	.217	.214	2.63	2.63
3446	11300	12230	6730	7740	3.86	4.69	2.34	2.35	1.57	2.01	.225	.167	2.58	2.50
3596	11380	11740	6900	7400	3.91	4.39	2.24	2.22	1.62	1.87	.209	.170	2.52	2.54
3675	13400	13710	7510	8660	5.18	6.30	3.77	3.16	2.04	2.70	.271	.168	2.68	2.67
3910	12200	10780	6940	6540	4.32	3.47	3.01	1.99	1.71	1.44	.261	.209	2.64	2.49
4020	11490	11670	6100	6970	3.45	4.07	2.93	2.45	1.32	1.66	.304	.223	2.64	2.54

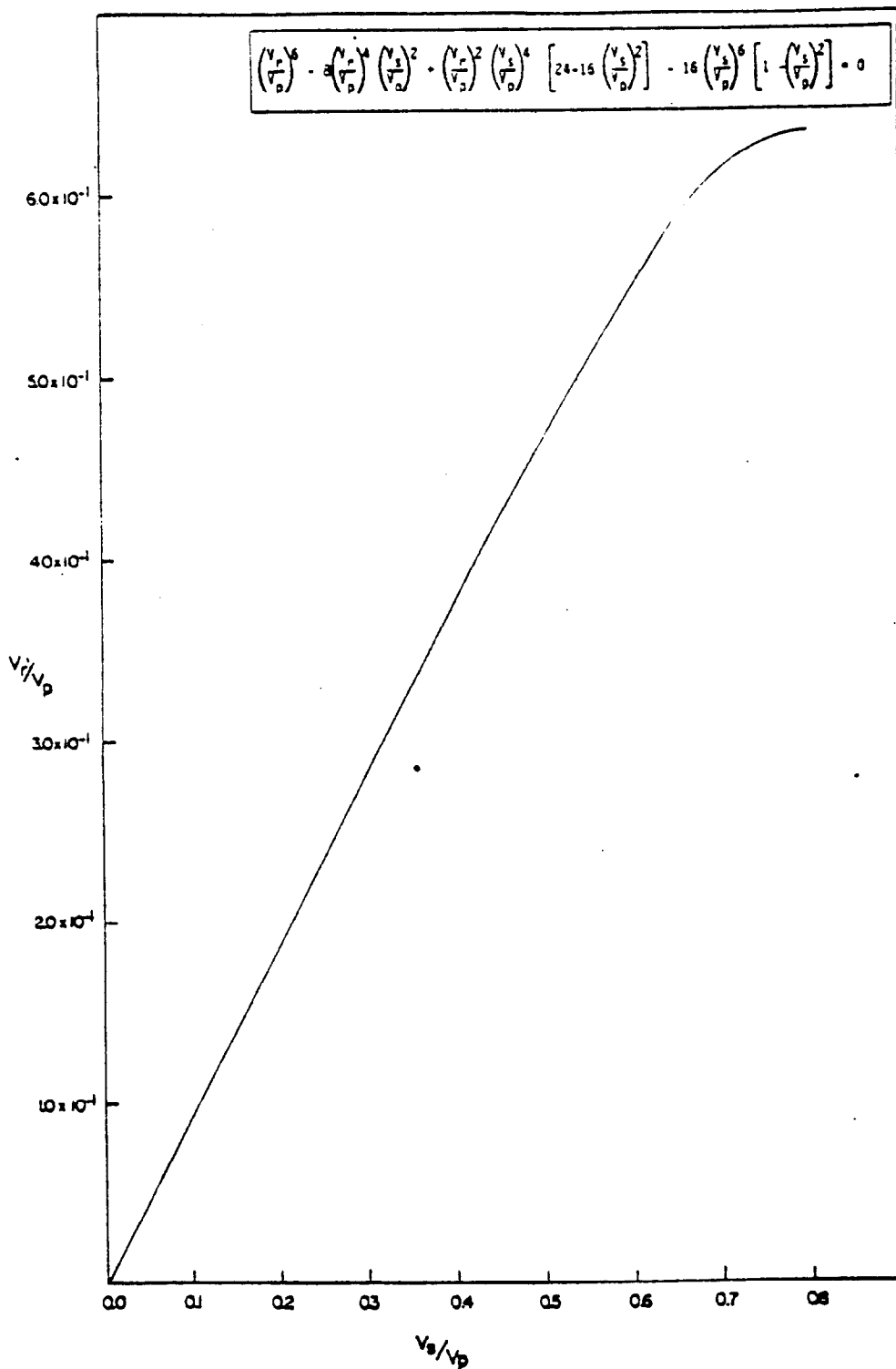


Figure 6. Plot of ratio of the Rayleigh wave ( $V_R$ ) velocity to the p-wave ( $V_P$ ) velocity as a function of the ratio of the s-wave ( $V_S$ ) velocity to the p-wave velocity.

TABLE III

P-wave, Rayleigh wave, and calculated s-wave taking the log s-wave data as the Rayleigh wave. From a Schlumberger experimental log.

Depth (ft)	Velocity (ft/sec)		
	Log $V_p$	Log $V_r$	Derived $V_s$
2880	13560	7630	8330
2891	13320	7260	7870
3110	12580	6800	7380
3365	11430	6870	7640
3446	11300	6730	7450
3596	13380	6900	7710
3675	13400	7510	8200
3910	12200	6940	7611
4020	11490	6100	6530

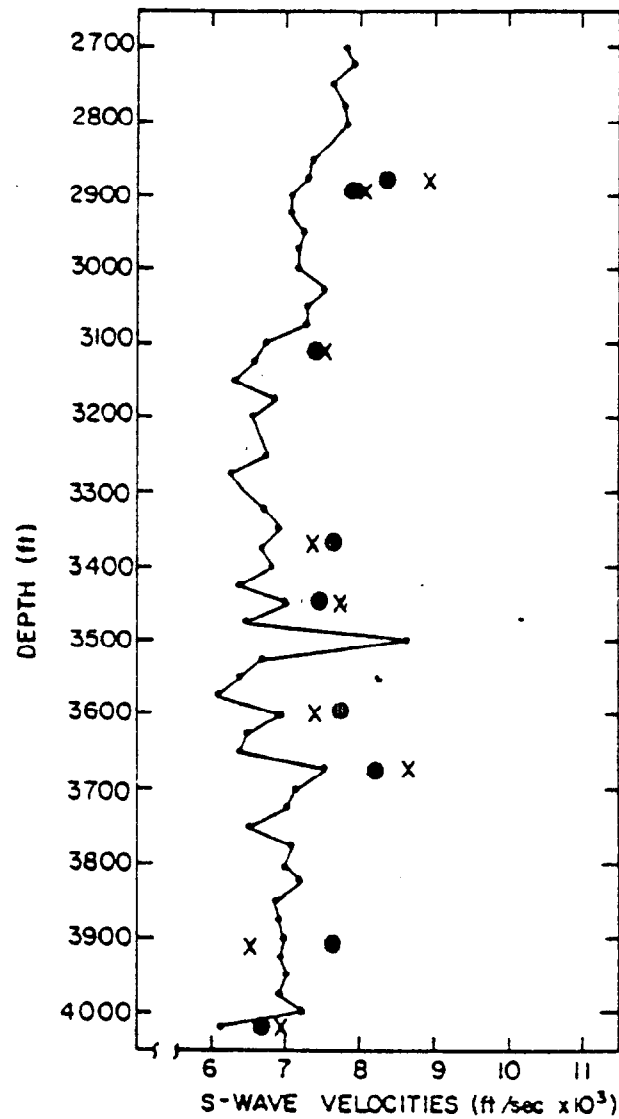


Figure 7. Comparison of log data, laboratory data (x) and calculated s-wave velocities from log data (•), assuming that the log detected the Rayleigh surface wave.

capabilities and limitations of a sonic logging tool as an elastic properties tool. Another project is presently under way to make similar comparisons on a well with considerably different lithology, in the Rocky Mountain area.

#### ACKNOWLEDGMENTS

This work was done by Terra Tek, Inc. in conjunction with Columbia Gas System Service Corporation under a U.S. Energy Research and Development Administration contract no. E(46-1)-8016. The authors wish to acknowledge the cooperation received from Dr. D. Rader, Schlumberger-Doll Research, who first pointed out the possibility of the logs detecting the Rayleigh wave. Appreciation is also expressed for the support received by Dr. A. Abou-Sayed, H. Morales, and K. Bradshaw of Terra Tek, Inc.

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#### APPENDIX A-1

##### ELASTIC MODULI FROM THE THEORY OF ELASTICITY FOR HOMOGENEOUS, ISOTROPIC SOLIDS

The elastic moduli and Poisson's ratio are obtained from the longitudinal velocity ( $V_p$ ), shear velocity ( $V_s$ ) and density ( $\rho$ ) of homogeneous, isotropic solids as follows:

$$\text{Poisson's Ratio, } \nu = \frac{V_p^2 - 2V_s^2}{2(V_p^2 - V_s^2)},$$

$$\text{Shear Modulus, } G = \rho V_s^2,$$

$$\text{Young's Modulus, } E = 3G \left[ \frac{V_p^2 - 4/3 V_s^2}{V_p^2 - V_s^2} \right],$$

$$\text{Bulk Modulus, } K = \rho(V_p^2 - 4/3 V_s^2).$$

APPENDIX B  
COMPARISON OF LABORATORY AND  
LOG DATA

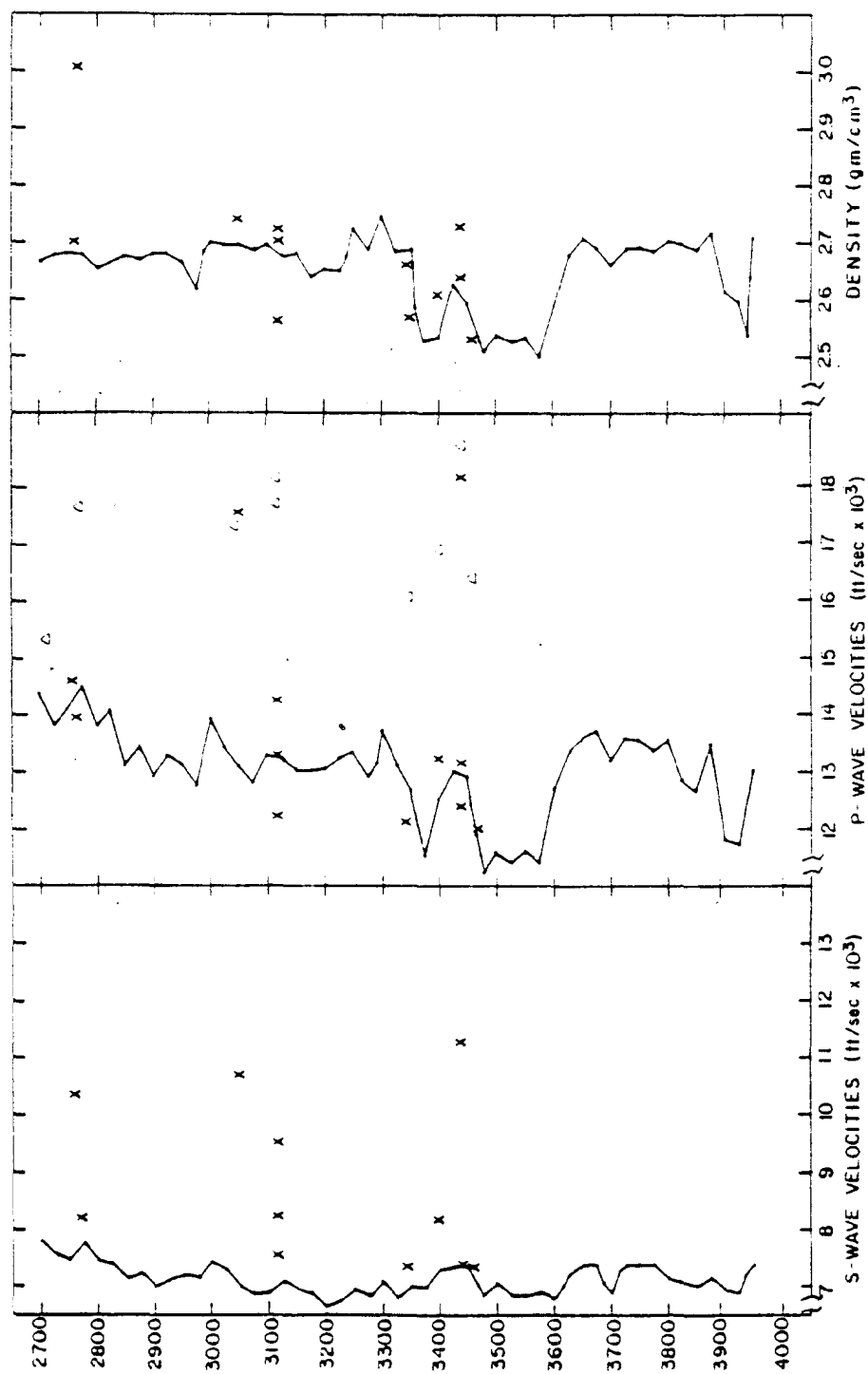


Figure B-1. Comparison of laboratory data with data from the Birdwell (3-D) log on Columbia Gas System Service Corp., well #20402 ("x" signifies velocity measurement made in a plane perpendicular to the borehole).



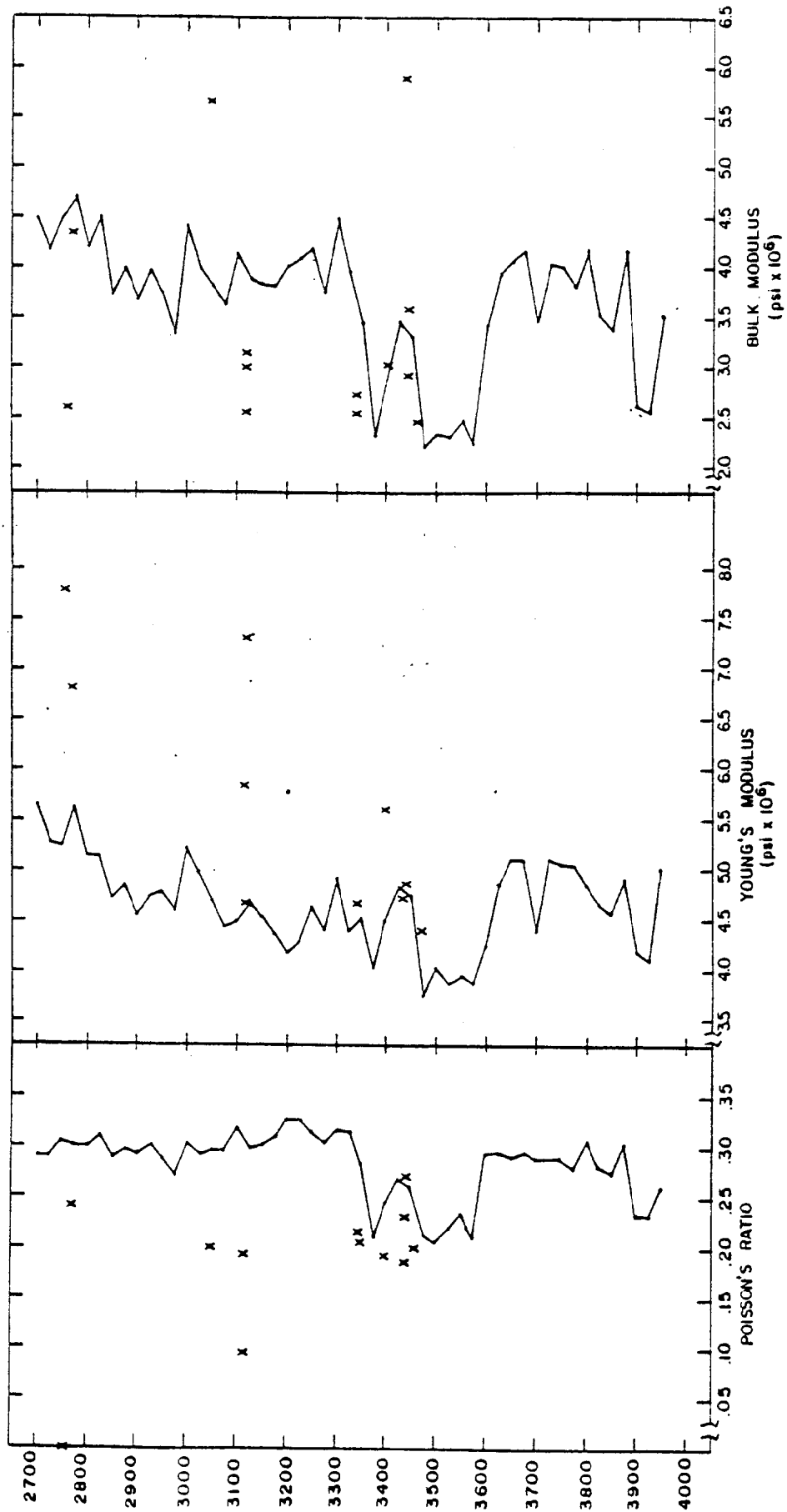


Figure B-2 Comparison of elastic properties obtained from laboratory and Birdwell (3-D) log data on Columbia Gas System Service Corp., well #20402.

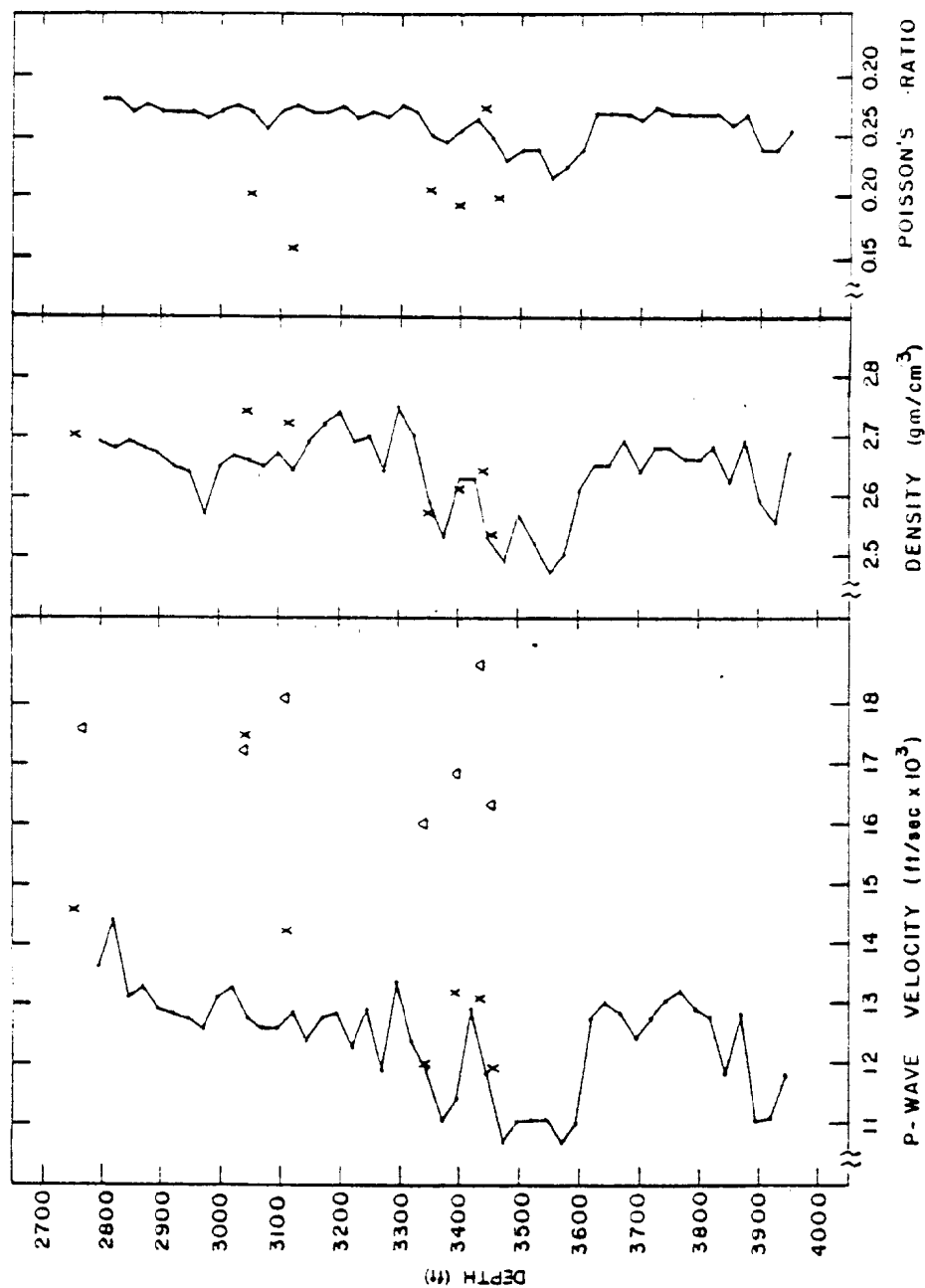


Figure B-3 Comparison of laboratory data with data from Schlumberger logs on Columbia Gas System Service Corp., well #20402. ("x" signifies velocity measurements made in a plane perpendicular to the borehole).

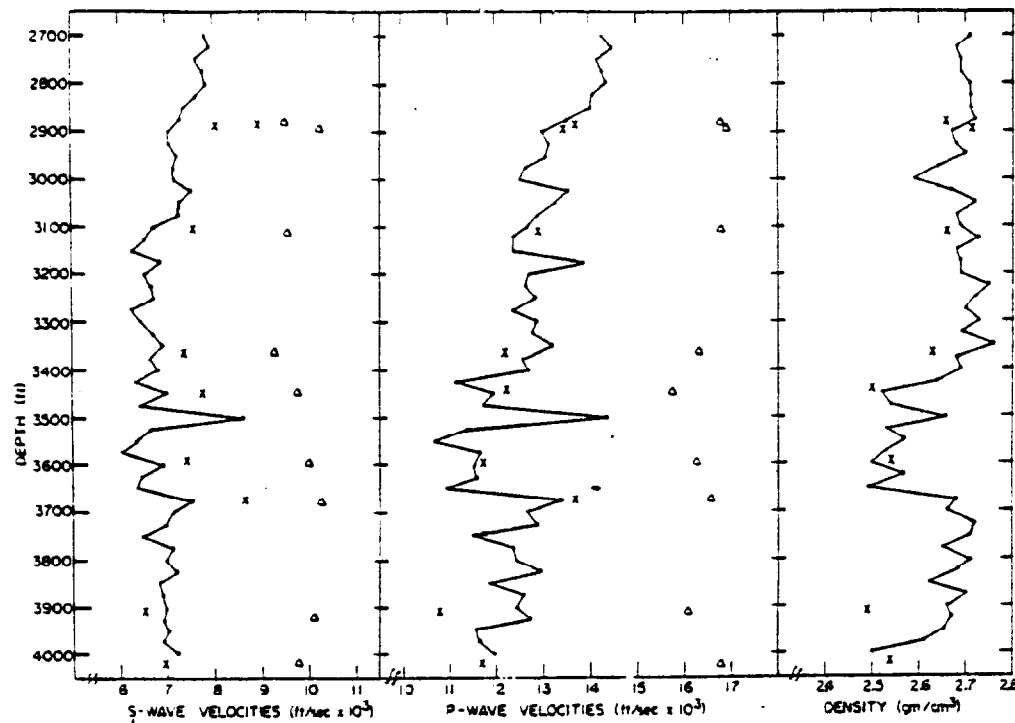


Figure B-4 Comparison of laboratory data (Terra Tek, Inc.) with data from a Schlumberger experimental s-wave logging tool on Columbia Gas System Service Corporation, well #20403. "X" indicates sample orientation parallel to borehole, "Δ" - perpendicular.

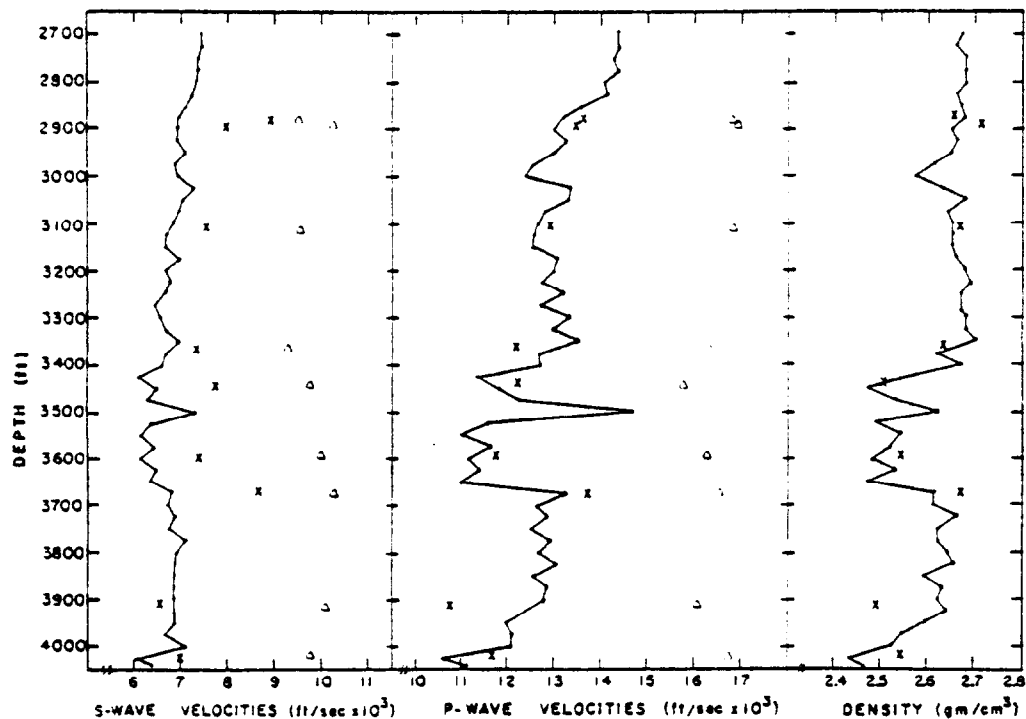


Figure B-5 Comparison of laboratory data (Terra Tek, Inc.) with data from the Birdwell (3-D) log on Columbia Gas System Service Corporation, well #20403. "X" indicates sample orientation parallel to borehole, "Δ" - perpendicular.

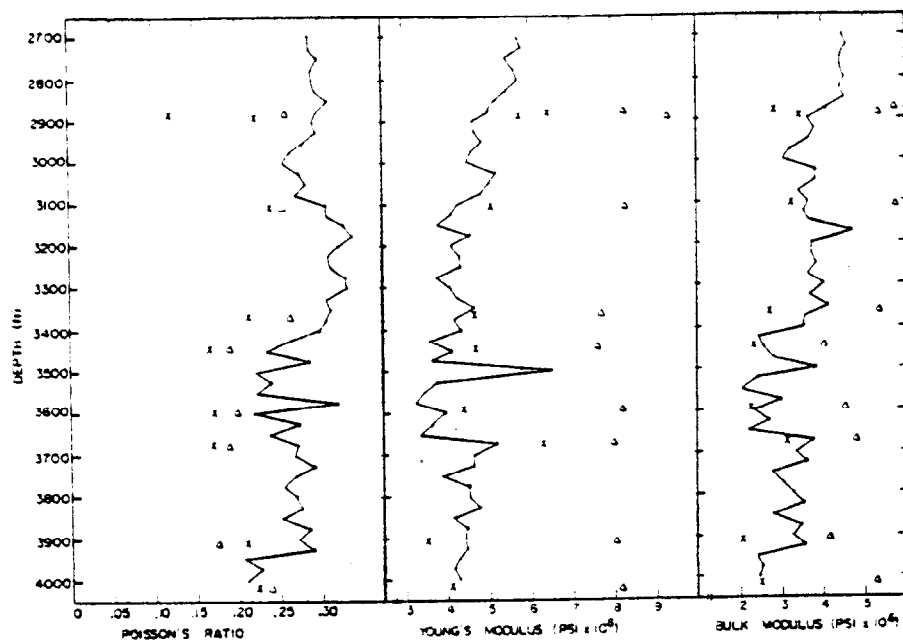


Figure B-6 Comparison of elastic properties derived from laboratory and log (Schlumberger) measurements on Columbia Gas System Service Corporation, well #20403. "X" indicates that the values were calculated from measurements on samples oriented parallel to the borehole, "Δ" - perpendicular.

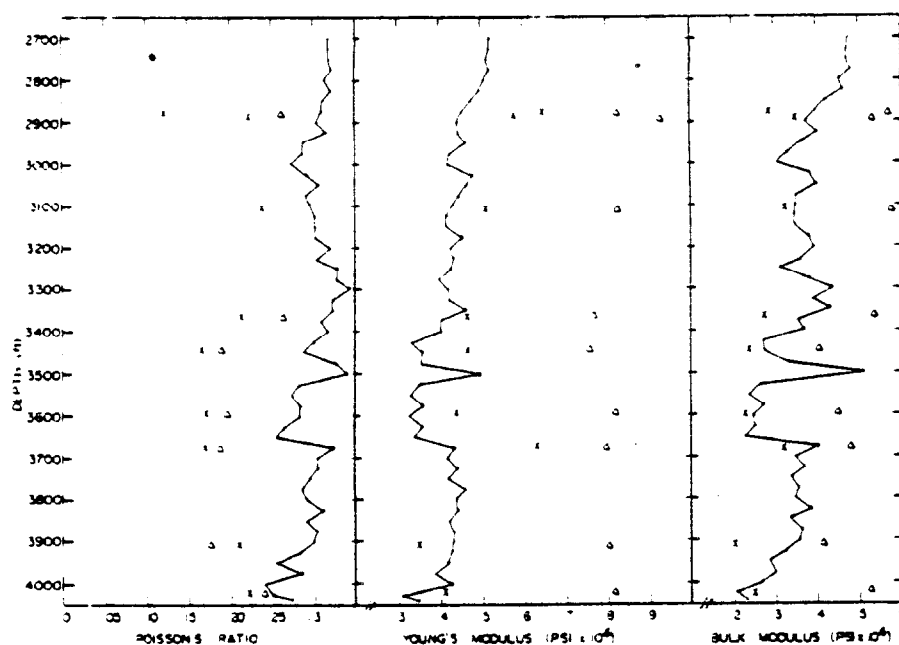


Figure B-7 Comparison of elastic properties calculated from laboratory measurements and data from the Birdwell (3-D) log on Columbia Gas System Service Corporation, well #20403. "X" indicates that the laboratory measurements were made on samples oriented parallel to the borehole, "Δ" - perpendicular.

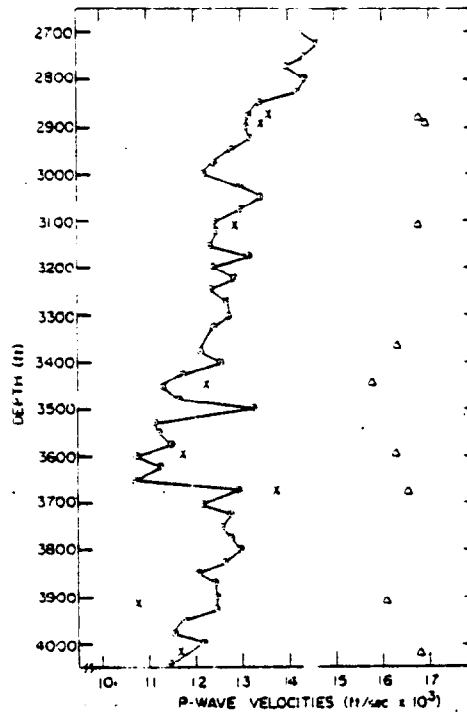


Figure B-8 Comparison of laboratory data with borehole compensated sonic log (Schlumberger) on Columbia Gas System Service Corporation, well #20403. "X" indicates samples oriented parallel to borehole, "Δ" - perpendicular

APPENDIX C  
METHOD FOR DETERMINING THE  
PERMEABILITY OF ROCK SAMPLES

Permeability can be measured or estimated by any one of several different methods. Typical laboratory determinations of permeability, however, are usually made using one of the following two techniques:

1. The constant flow or steady-state method which uses a flow meter or positive displacement pump to measure or control the flow rate through the sample. This method requires the flow rate through the material and the pressure drop across the material to be constant at the time of measurement.
2. The transient method imposes a step increase in pressure in a known volume across a sample. The permeability can be calculated from the time-dependent decay of this imposed pressure step.

Each method has advantages and disadvantages over the other depending on the conditions required for the test and the permeability of the sample in question. The first method is generally used for porous media having permeabilities greater than 100  $\mu$ darcies, while the second method is more adaptable to low porosity materials such as tight sandstones where the permeabilities are in the tens of microdarcies or lower.

A sketch of a typical test set-up using the transient method for measuring permeability is shown in Figure C1. The sample is placed in a pressure vessel and the pore-pressure inlet and outlet lines are connected to external fittings through the base plug. With this geometry, the sample can be subjected to hydrostatic loading or triaxial compression prior to testing. Pore pressure in the sample can be set at any value less than the confining pressure.

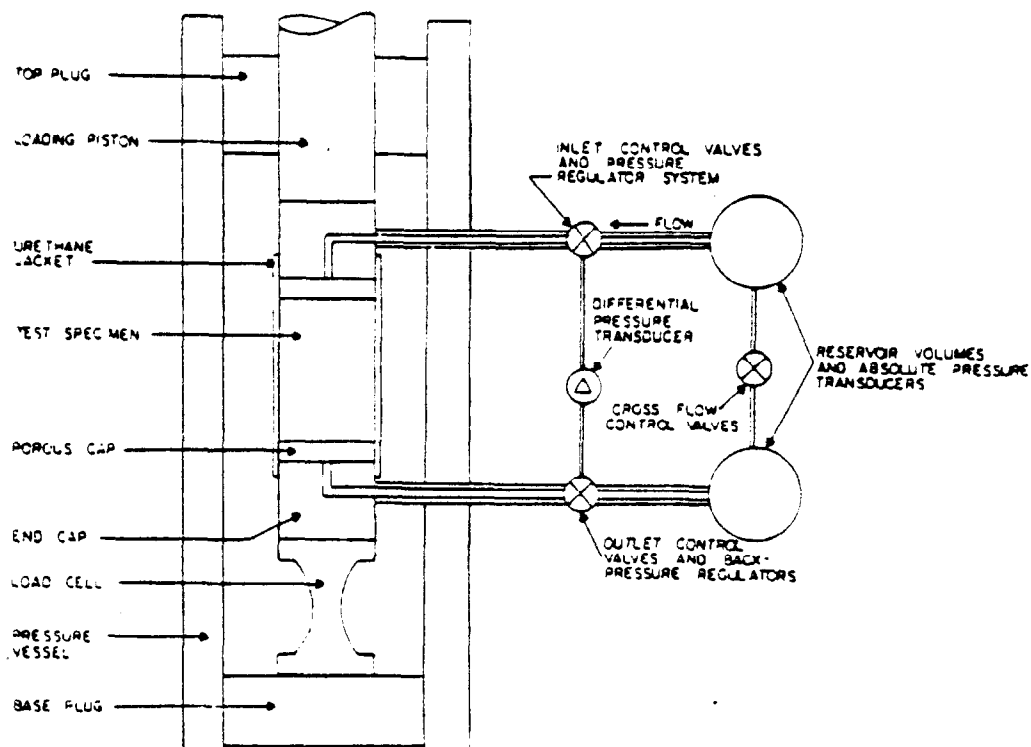


Figure C1. Schematic drawing of the test apparatus used to measure permeability.

Figure C2 illustrates the volumes of fluids on either side of the sample that can be hydraulically connected to allow the pore pressure to equalize. When the sample has reached equilibrium, the volumes are disconnected by closing a valve. The pressure in volume one is then raised slightly. This pressure step should be limited to a small percentage (less than 5 percent) of the absolute pressure in the reservoir volumes. The pressure step decay is monitored accurately through the use of a differential pressure transducer. Variations of sample length and volume of the reservoirs can be changed to allow the test to be completed in a convenient length of time. A brief outline of the theory involved in measuring permeabilities using the transient technique is given below. A detailed treatment of this analysis is presented by Brace, *et al.*, (1968).



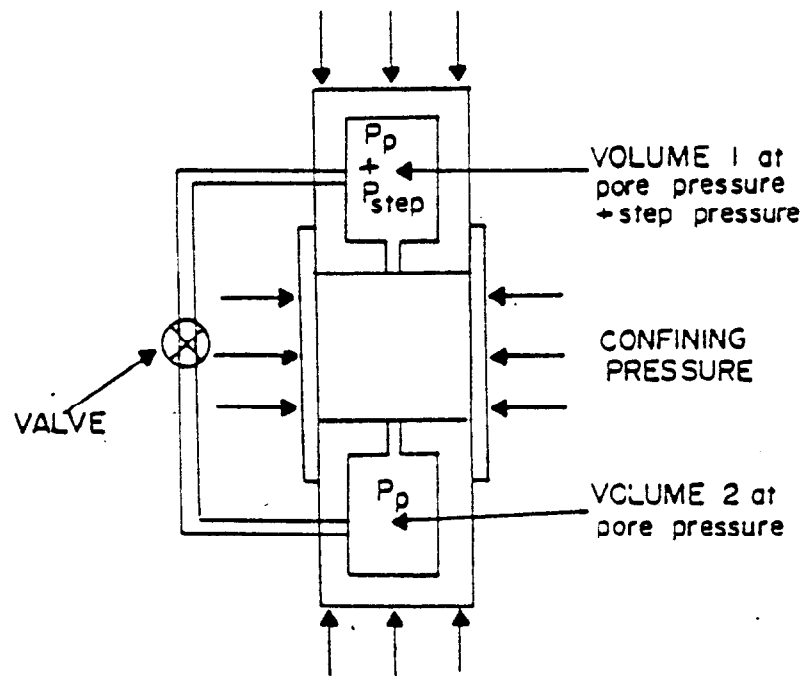


Figure C2. Schematic drawing of a permeability test using the transient technique.

The equation for compressible flow in a compressible media is

$$\nabla^2 p = \frac{\mu \beta}{k} C \frac{\partial p}{\partial t} \quad (1)$$

Where

$\mu$  = fluid viscosity

$\beta$  = fluid compressibility

$k$  = permeability

$C$  = a term which includes the compressibility of the rock matrix

and

$p$  = pressure

Equation (1) can be solved under the following assumptions concerning fluid flow characteristics:

1. Darcy's law is valid.
2. The fluid flow is laminar.
3. The change in fluid volume in the pores of the rock due to the step pressure change is negligible compared to the amount of fluid flowing through the sample during a test.
4. The pressure step is small compared to the absolute pore pressure so that the physical constants of the fluid (viscosity and compressibility) can be considered constant in all parts of the sample.

Under these conditions the solution to Equation (1) is given by

$$\Delta P = \overline{\Delta P} \frac{V_2}{V_1 + V_2} e^{-\alpha t} \quad (2)$$

where

$\overline{\Delta P}$  = Initial pressure step

$\Delta P$  = (Instantaneous pressure) - (Final pressure), i.e.,  $(P - P_f)$

$V$  = Volume of reservoir at either end of the sample

The permeability  $k$  is given by the equation as

$$k = \frac{\alpha \beta \mu l}{A(1/V_1 - 1/V_2)} \quad (3)$$

where

$\alpha$  = the slope of the semilog of the natural log of the decaying pressure versus time (as defined in Equation 2)

$l$  = the sample length

$A$  = the sample area

Thus, the permeability can be accurately determined for very tight samples with no direct measurement of the flow rate. Another major advantage is the capability for making permeability measurements at high pore pressures.

APPENDIX D  
FLOW CAPACITY MEASUREMENTS

The flow capacity is reported rather than the permeability because the fracture width is generally not known in the tests.

The flow capacity of the fracture is determined from the equation

$$kw = \frac{2000 Q_o p_o L \mu}{(p_i^2 - p_o^2) h} \quad [\text{md-cm}]$$

where

- k = Permeability (millidarcy's)
- $Q_o$  = Flow Rate of Outlet Air ( $\text{cm}^3/\text{sec}$ )
- $p_o$  = Outlet Pressure (Atm absolute)
- $p_i$  = Inlet Pressure (Atm absolute)
- $\mu$  = Viscosity (centipoises)
- L = Length of Sample (cm)
- w = Width of Fracture (cm)
- h = Height of Fracture (cm).

In the test nitrogen flows in and out of the pressure vessel through small lines with resulting pressure losses; therefore, a second set of lines were used to sense gas pressures at the ends of the samples. In this way, and for steady state flow, pressures were measured directly and no corrections are needed for line losses. The volume of gas flowed were measured at atmospheric pressure at the end of small flow lines leading from the pressure vessel. The volumetric flow at the outlet end of the sample was determined by making the pressure correction between the flow meter atmospheric pressure and the sample outlet pressure assuming isothermal flow. Figure D1 is a schematic diagram of the experimental set-up.

For the unpropped sample the flow was low enough to be within the measurement range of the equipment used in the transient method, so several tests were made by this technique to cross-check the steady state technique. This data is also given in Table DI and plotted in Figure D2.

After completion of the unpropped flow tests, the sample faces were viewed under a low-power (3x) stereo microscope. The contact points were essentially a discoloration of the surface with little or no visible crushing or deformation. The observations indicate that the points of contact were highly non-uniform. Over a large portion of the faces, perhaps only 5 to 15 percent of the area was marked. In some spots perhaps 60 to 80 percent of the surface was marked. Figure D3 is a photograph of the sample. No attempt was made to quantitatively measure either the surface roughness or the fraction of the surfaces that appeared to make contact when the two halves were fitted back together for the test.

A rough estimate of the effective width of the flow channels in the unpropped fracture can be made assuming equivalent permeability for flow between parallel plates (Craft and Hawkins, 1959).

$$k[\text{darcys}] = 54 \times 10^6 w^2 [\text{inches}^2]$$

This relationship needs to be modified to account for the rough surface and contact points that hold the crack open. One possible modification might be adapted from Carmen (1970) to give

$$k = \frac{8.38 \times 10^6 w^2 \phi^3}{T(1 - \phi^2)}$$

where

$k$  = permeability in millidarcies

$\phi$  = fraction of surface open to flow

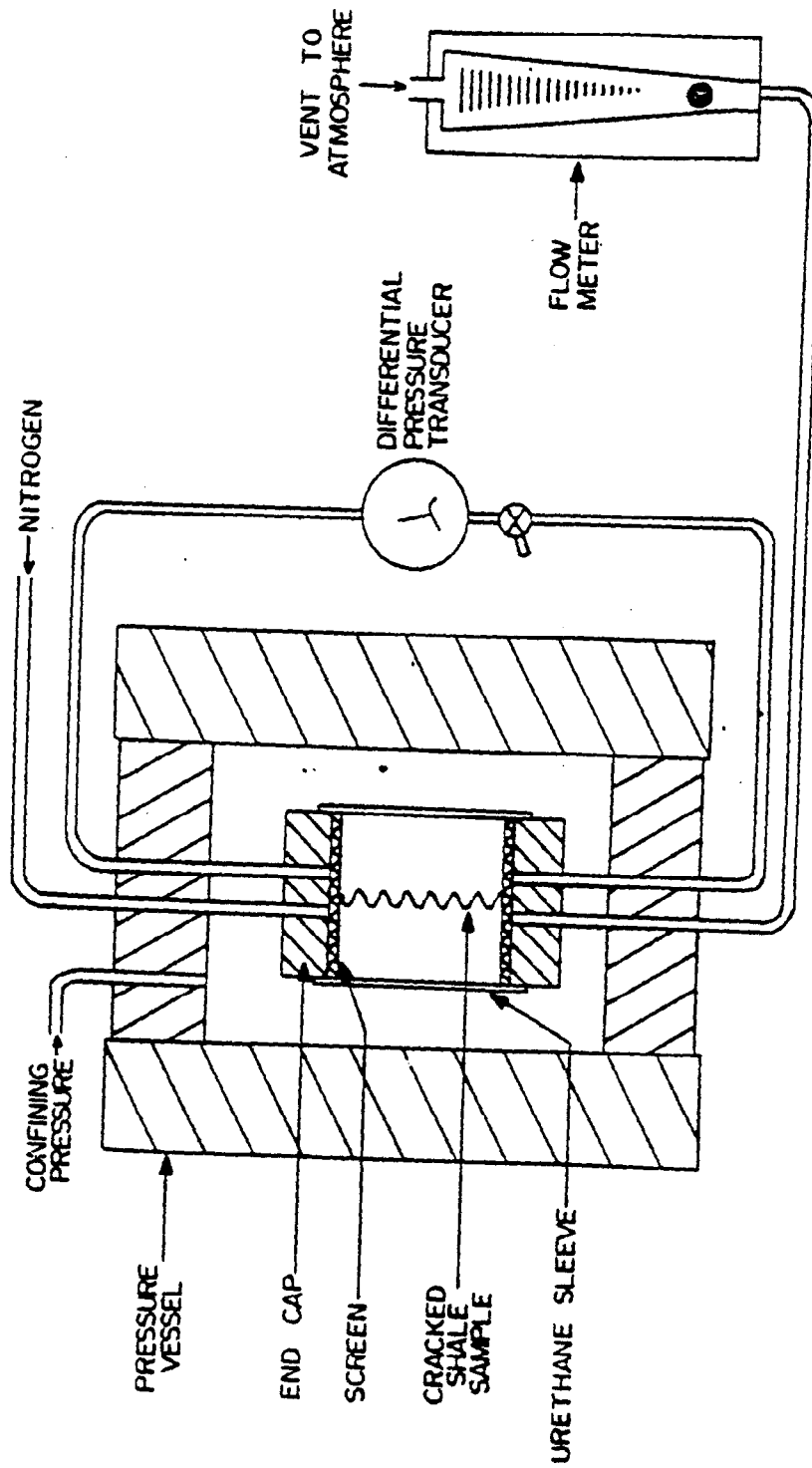


Figure D1. Schematic design of flow test set-up

TABLE DI  
Data for Flow Through Unpropped  
Fracture in Shale\*

First Steady State Test					
Confining psi	Q ml/sec	P <sub>i</sub> atm	P <sub>o</sub> atm	1/ $\bar{P}$	kw md-cm
100	19.7	3.034	3.009	.165	9900
1000	17.0	9.1	8.4	.057	300
2000	41.1	10.66	1.44	.083	13.5
2900	13.8	20.5	4.27	.040	3.8
4000	17.3	24.9	1.36	.038	.97
Transient test					
2900					7.1
4000					3.4
4000					2.6
Second Steady State Test					
100	1.33	.8574	.8571	1.166	56,500
	4.38	.8660	"	1.161	6,240
	7.08	.8728	"	1.156	5,690
	20.58	.9027	"	1.137	5,600
	15.0	.8837	"	1.149	7,020
	79.2	1.0415	"	1.053	4,940
	170.0	1.3475	"	1.907	3,430
1000	1.75	.9150	.8571	1.129	370
	4.55	.8769	"	1.153	2,890
	7.17	1.1782	"	.983	240
	15.33	1.4837	"	.854	228
	11.25	1.2701	"	.940	280
	18.17	1.5707	"	.824	229
2000	1.5	1.0687	.8571	1.039	80.4
	4.48	1.5401	"	.834	59.3
	3.75	1.5095	"	.845	53.0
4000	.65	1.2293	.8571	.959	18.3
	1.92	1.5912	"	.817	23.3
	4.33	3.1755	"	.496	10.1
	4.42	3.4721	"	.462	8.5
	16.67	7.8231	"	.230	6.0
	15.00	6.7415	"	.253	7.3

\* Flow with dry nitrogen

$\mu$  = .018 cp

$h$  = 8.25 cm

$L$  = 5.84 cm



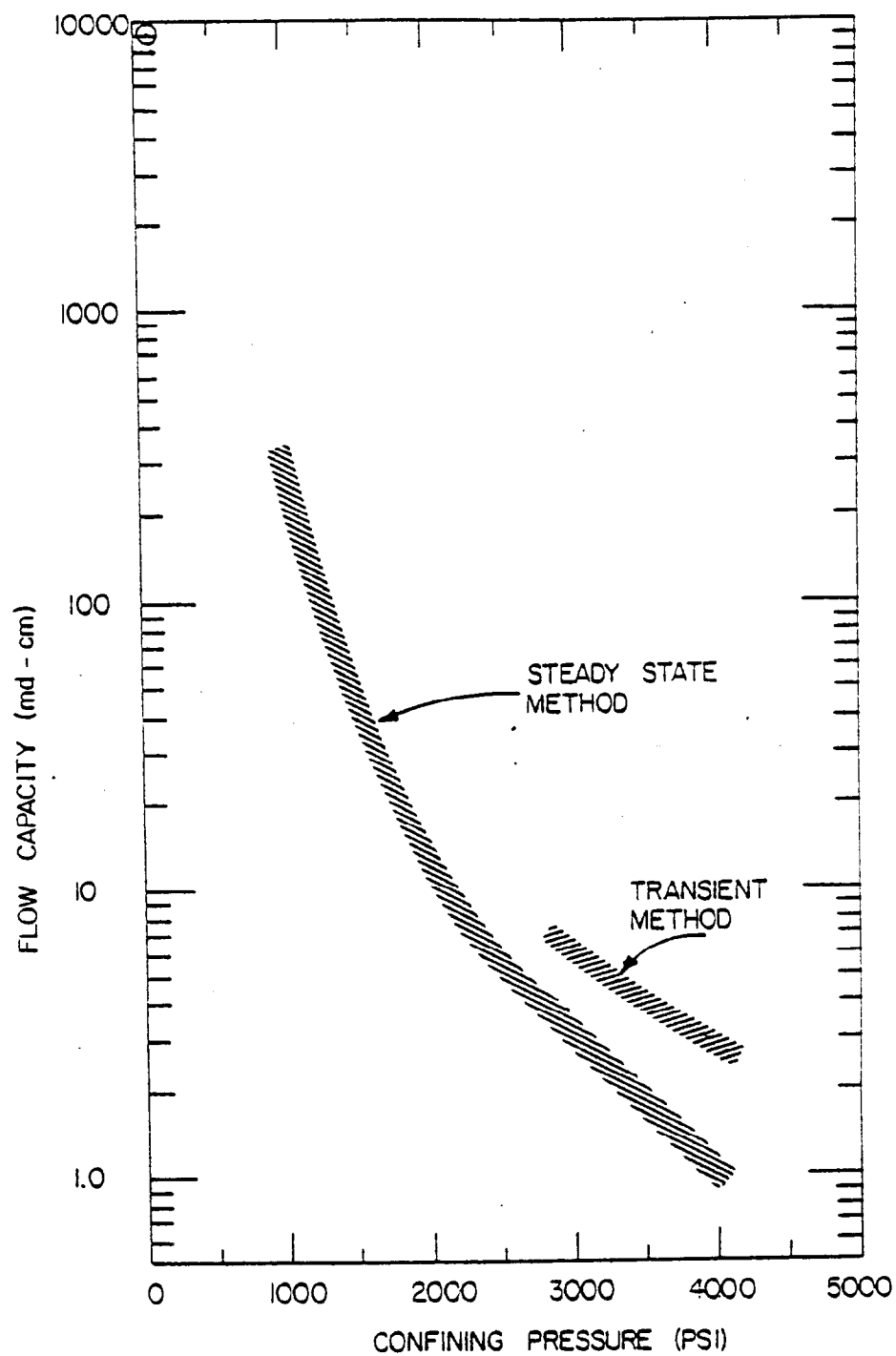


Figure D2. Flow through the unpropped fracture.

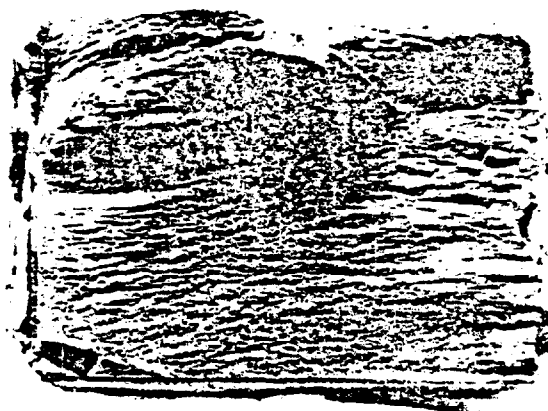


Figure D3. Photograph of test sample

$t$  = tortuosity (actual channel length/straight line)

$w$  = crack width in cm

Multiplying by  $w$  and rearranging to solve for  $w$  in terms of the flow capacity ( $kw$ ) gives the equation

$$w \approx \left[ \frac{0.12 \times 10^{-6} T (1-\phi^2) kw}{\phi^3} \right]^{1/3}$$

It is evident from the data that the flow capacity is significantly dependent on the gas pressure. This is assumed to be the Klinkenberg effect of gas slippage. The pressure-dependent permeability is related to the absolute permeability by the Klinkenberg equation

$$k = k_0 (1 + b/\bar{P})$$

where

$k_o$  = absolute permeability

$\bar{P}$  = mean pressure in sample

$b$  = Klinkenberg constant

In a plot of  $k_w$  versus  $1/\bar{P}$ , the intercept is the absolute flow capacity  $k_o w$  and the slope of the line is equal to  $k_o w b$  with  $\bar{P} = (P_i + P_o)/2$  and the absolute flow capacity is given by

$$k_o w = \frac{2000 Q_o \mu L}{(P_i + P_o + 2b) (P_i - P_o) h}$$

TERRA TEK REPORT

FRACTURE FLOW CAPACITY OF  
HYDRAULICALLY FRACTURED DEVONIAN SHALES

FRACTURE FLOW CAPACITY OF HYDRAULICALLY  
FRACTURED DEVONIAN SHALES

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## ABSTRACT

Flow capacities were determined for induced fractures in cores taken from 3445 feet and 3695 feet in the Columbia Gas System Services Corporation Well #20403, located in Lincoln County, West Virginia. The samples from depth 3445 feet were from the 'Middle Brown Shale' and from depth 3695 feet the 'Lower Gray Shale,'. The work was aimed at assessing flow capacity damage potential of a number of water-based fracturing fluids. The fractures were propped with a partial monolayer ( $0.027 \text{ lb/ft}^2$ ) of 20/40 mesh sand.

At conditions simulating *in situ* closure stress (2700 psi) and temperature (70°F), the 'Middle Brown Shale' fracture flow capacity was reduced to 5 percent of the original flow capacity. For the 'Lower Gray Shale' (the *in situ* closure stress of 2900 psi, temperature 70°F) the reduced flow capacity was close to one percent of the original. In both shales the decrease in flow capacity resulted from sand embedment initiated by fracturing fluid softening of the rock as well as clay flocculation around the imbedded sand.

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## INTRODUCTION

Degradation of matrix and fracture permeability due to the application of hydraulic fracturing fluid has been presented as one reason for the failure of massive hydraulic fractures (Davis, 1974; Clark, 1977). The selection of a fracturing fluid is not only dependent upon the fluids effectiveness in creating the fracture and transporting the proppants; it is also dependent on the degree of formation damage and plugging. In a recent study (Holditch, 1978) the overall productivity decrease in gas production from the combined effects of reservoir damage, relative permeability damage, capillary pressure damage and fracture conductivity damage were investigated. Reduction in fracture conductivity had significant effect on productivity. Thus, the necessity of experimentally determining the damaging effect of fracturing fluid to the flow capacity of the specific formation.

## EXPERIMENTAL PROCEDURE

Core samples taken from Columbia Gas System Services Corporation Well #20403 at depths of 3445 feet and 3695 feet were used in this investigation. The samples for the 3445 feet depth were from the 'Middle Brown Shale' and 3695 feet depth from the 'Lower Gray Shale'. The work was aimed at assessing flow capacity damage of a number of water-based fracturing fluids to fractures propped with a partial monolayer of 20/40 mesh sand.

The core samples were saw cut and propped with a sand concentration of 0.027 lb/ft<sup>2</sup>. Initially the cores were subjected to confining pressure of 90 psi for the proppants to settle in place. By flowing dry nitrogen gas through the propped channel, flow capacity measurements were taken. The change in flow capacity with effective pressure was determined by varying the confining pressure from 500 psi to 3500 psi; in all cases gas pressure within the fracture was maintained at 300 psi. Cantilevers were placed on the outer core surface to monitor changes in fracture width closure. Fracturing fluid was subsequently flowed through the propped fracture for four hours (to simulate field fracturing time) and the change in flow capacity with effective pressure was determined for the same confining pressure range.

The constituents of the fracturing fluids and the test conditions are presented in Table 1 and 2 respectively. All fracturing fluids were supplied by Dowell. Besides the fracturing fluids saturated nitrogen was flowed through the propped fracture to assess the fracture flow damage from water alone. After each sequence of tests the fractures were examined with an optical microscope to assess the degree of sand embedment, sand crushing, and clay flocculation in the fracture.

TABLE 1  
FRACTURING FLUIDS

SYSTEM	CONSTITUENTS
1. Waterfrac 20-40	20-40 #J266/1000 gal + 1 gal F75N/1000 gal w/fresh water
2. Waterfrac 20 W/CO <sub>2</sub>	20 #J160/1000 gal + 10 gal L42/1000 gal + 1 gal F75N/1000 gal w/fresh water
3. Superfoam	8 gal W22/1000 gal + 10 gal L42/1000 gal w/fresh water

TABLE 2  
TEST CONDITIONS

Confining Pressure	500 - 3500 psi
Fluid Pressure within the Fracture	300 psi
Temperature	70°F
Flowing Fluid	Dry Nitrogen
Fracturing Fluid Injection Pressure	850 psi

## DISCUSSION OF RESULTS

Discussion of results are presented in light of the type of shales.

*Middle Brown Shale:* Figure 1 and Table 3 illustrate the change in the fracture flow capacity with increasing effective pressure for the virgin sample and after exposure to Waterfrac 20 W/CO<sub>2</sub> and saturated nitrogen. Figure 2 and Table 4 show the decreasing trend of the calculated effective fracture width with the increase in effective pressure for the same tests. The gentle slope of the curves for the virgin sample in both Figure 1 and 2 suggests that the fracture closed mainly as a result of proppant embedment. Figure 2 also includes a plot of the fracture width (derived from experimentally measuring the closure width) with effective pressure for the virgin sample and upon being interacted by Waterfrac 20 W/CO<sub>2</sub>. This provides a qualitative and quantitative comparison between calculated and experimentally measured values.

Upon application of the fracturing fluid there is a marked reduction in fracture conductivity. Waterfrac 20 W/CO<sub>2</sub> fracturing fluid decreased the original flow capacity by approximately two orders of magnitude. Saturated nitrogen had an even greater effect on the flow capacity. This clearly explains the effect of water on the fracture surface.

Optical microscopic examination of the fracture face after interaction with Waterfrac 20 W/CO<sub>2</sub> is shown in Figure 4. Evidence of deep sand embedment is present with signs of clay flocculation around the proppants.

The following reasons can be accepted as causes for the overall decline in flow capacity due to fracturing fluid application:

TABLE 3  
COLUMBIA GAS SYSTEM WELL #20403  
MIDDLE BROWN SHALE  
3445'  
COMPARISON OF FRACTURE FLOW CAPACITY

Effective Pressure psi	Fracture Flow Capacity md-cm		
	Before Fracturing Fluid Flow	After Fracturing Fluid Flow	
		Waterfrac 20 W/CO <sub>2</sub>	Saturated Nitrogen
200	92,000	2900	875
500	78,000	2550	850
1000	67,000	2150	810
2000	44,000	1880	760
3000	30,500	1670	730
3200	26,750	1650	720

TABLE 4  
COLUMBIA GAS SYSTEM WELL #20403  
MIDDLE BROWN SHALE  
3445'  
COMPARISON OF FRACTURE WIDTH

Effective Pressure psi	Fracture Width cm		
	Before Fracturing Fluid Flow	After Fracturing Fluid Flow	
		Waterfrac 20 W/CO <sub>2</sub>	Saturated Nitrogen
500	.0220	.00720	.00440
1000	.0207	.00717	.00420
2000	.0190	.00690	.00392
3000	.0170	.00670	.00365
3200	.0155	.00668	.00360

1. The water in the water based fracturing fluids helped the fracture face to soften and result in sand proppant embedment.
2. From Figure 3 we have signs of material clusters only around the proppants and no damage to the surface where there were no proppants. This suggests the fracturing fluid has no chemical action; fracture surface damage beneath proppants indicates clay softening. There is no evidence of clay swelling.
3. Fracture flow capacity decrease is due to proppant embedment.

*Lower Gray Shale:* Figure 4 and Table 5 show the change in the fracture flow capacity with increasing effective pressure for the virgin sample and after exposure to Waterfrac 20-40, superfoam and saturated nitrogen. Figure 5 and Table 6 illustrate the decreasing trend of the calculated effective fracture width with the increase in effective pressure for the same tests. Similar to the 'Middle Brown Shale', Figure 4 and 5 suggest that the fracture of the virgin sample closed mainly as a result of sand proppant embedment. Figure 5 also includes a plot of the fracture width (derived from experimentally measuring the closure width) with effective pressure for the virgin sample and upon being interacted by Waterfrac 20-40.

Upon application of the fracturing fluid there is a marked reduction in fracture conductivity similar to that seen for the 'Middle Brown Shale'. Both the Waterfrac 20-40 and Superfoam decreased the virgin flow capacity by roughly three orders of magnitude. Waterfrac 20-40 causing slightly less damage than Superfoam. Saturated nitrogen decreased the virgin flow capacity between one and two orders of magnitude. This is less than the effect seen for 'Middle Brown Shale'. From two separate studies, Leventhal (1978) and Mcketta (1978) it has been identified that 'Middle Brown Shale' has a higher

TABLE 5  
COLUMBIA GAS SYSTEM WELL #20403  
LOWER GRAY SHALE  
3695'  
COMPARISON OF FRACTURE FLOW CAPACITY

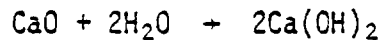
Effective Pressure psi.	Fracture Flow Capacity md-cm			
	Before Fracturing Fluid Flow	After Fracturing Fluid Flow		
		Waterfrac 20-40	Saturated Nitrogen	Superfoam
200	88,000	380	3950	140
500	75,000	220	3600	105
1000	63,000	130	3200	77
2000	41,000	98	2475	52
3000	26,500	74	1910	35
3200	24,750	69	1800	33

TABLE 6  
COLUMBIA GAS SYSTEM WELL #20403  
LOWER GRAY SHALE  
3695'  
COMPARISON OF FRACTURE WIDTH

Effective Pressure psi	Fracture width cm		
	Before Fracturing Fluid Flow	After Fracturing Fluid Flow	
		Waterfrac 20-40	Superfoam
500	.0210	.0036	.00256
1000	.0205	.0026	.00227
2000	.01805	.00234	.00193
3000	.0160	.00210	.00165
3200	.0147	.00201	.00161



percentage of organic materials and calcium oxide (CaO). Organic material absorbs water and calcium oxide absorbs water by chemically reacting with water in the following manner:



This explains why the 'Middle Brown Shale' has a lower flow capacity than 'Lower Gray Shale' upon being interacted by saturated nitrogen.

Optical microscopic examination of the fracture face after interaction with Waterfrac 20-40 and Superfoam are shown in Figure 6 and 7 respectively. Evidence of deep sand proppant embedment is present with signs of clay flocculation around the proppants. Fracture face interacted by Superfoam has more flocculated clay.

Reasons for the reduction of flow capacity in the 'Lower Gray Shale' due to the interaction by the fracturing fluids are the same as for the 'Middle Brown Shale'.

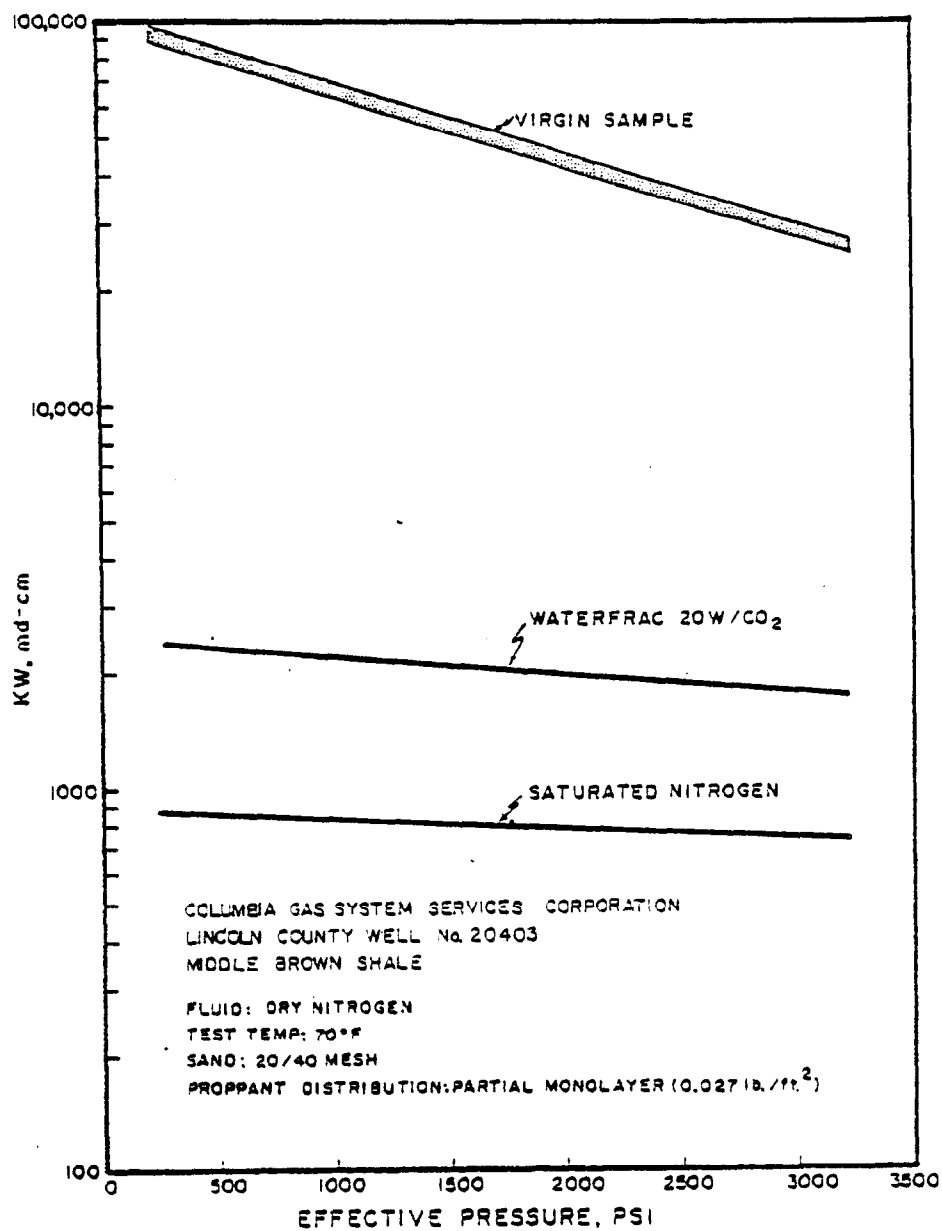


Figure 1. Trend of fracture flow capacity with the increase in effective pressure for 'Middle Brown Shale'.

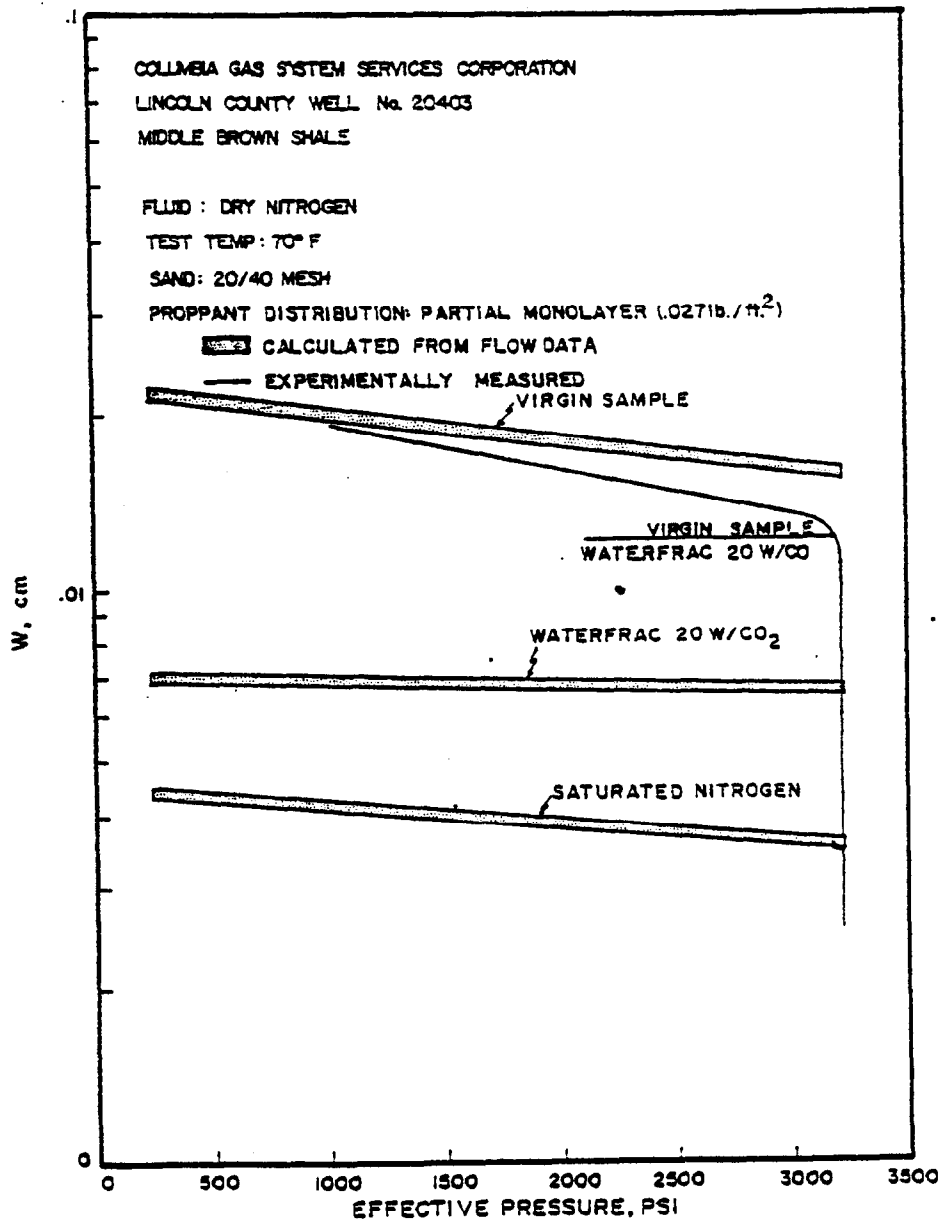


Figure 2. Trend of the effective fracture width with the increase in effective pressure for 'Middle Brown Shale'.

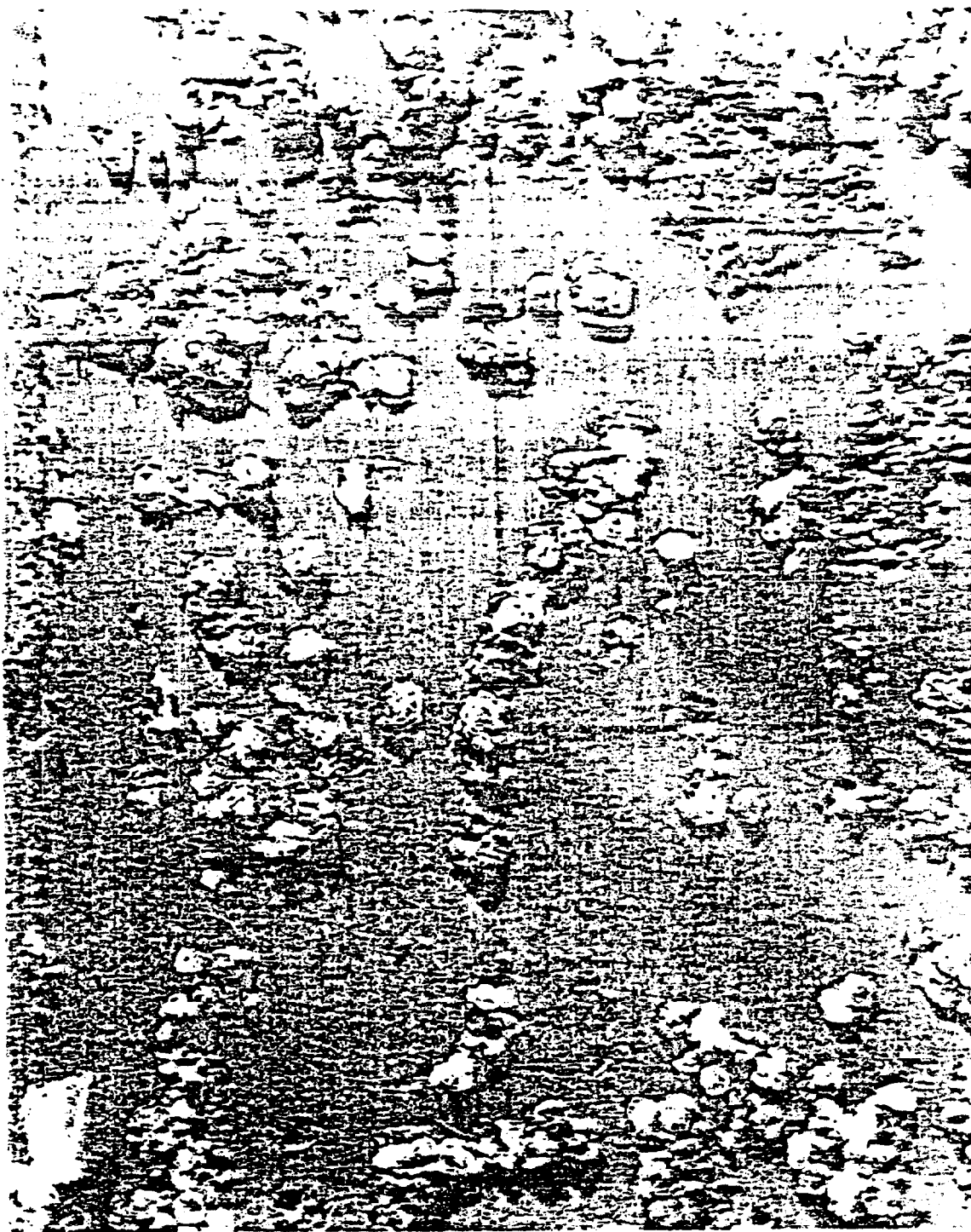


Figure 3. Fracture face of the 'Middle Brown Shale' sample interacted by Waterfrac 20 W/CO<sub>2</sub>.

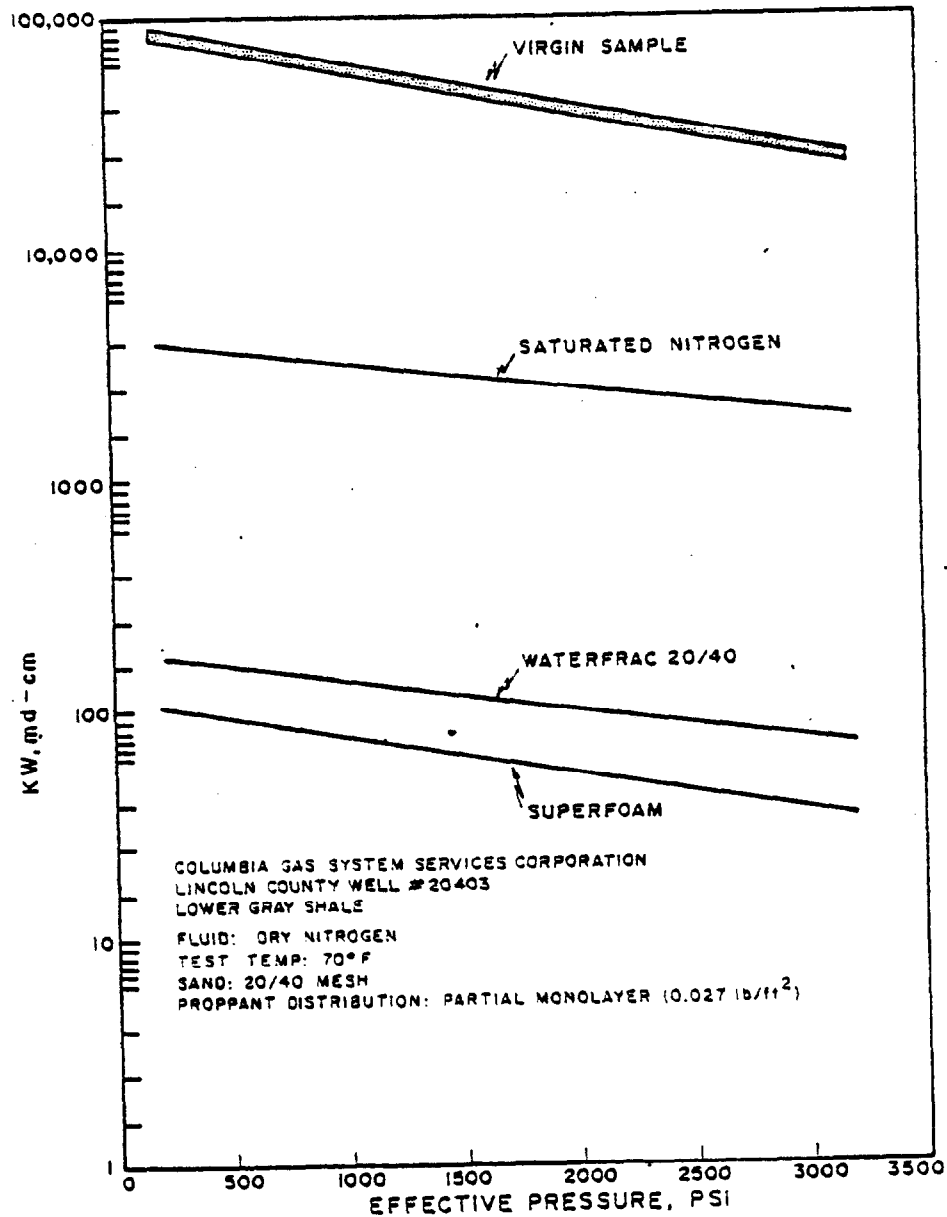


Figure 4. Trend of fracture flow capacity with the increase in effective pressure for 'Lower Gray Shale'.

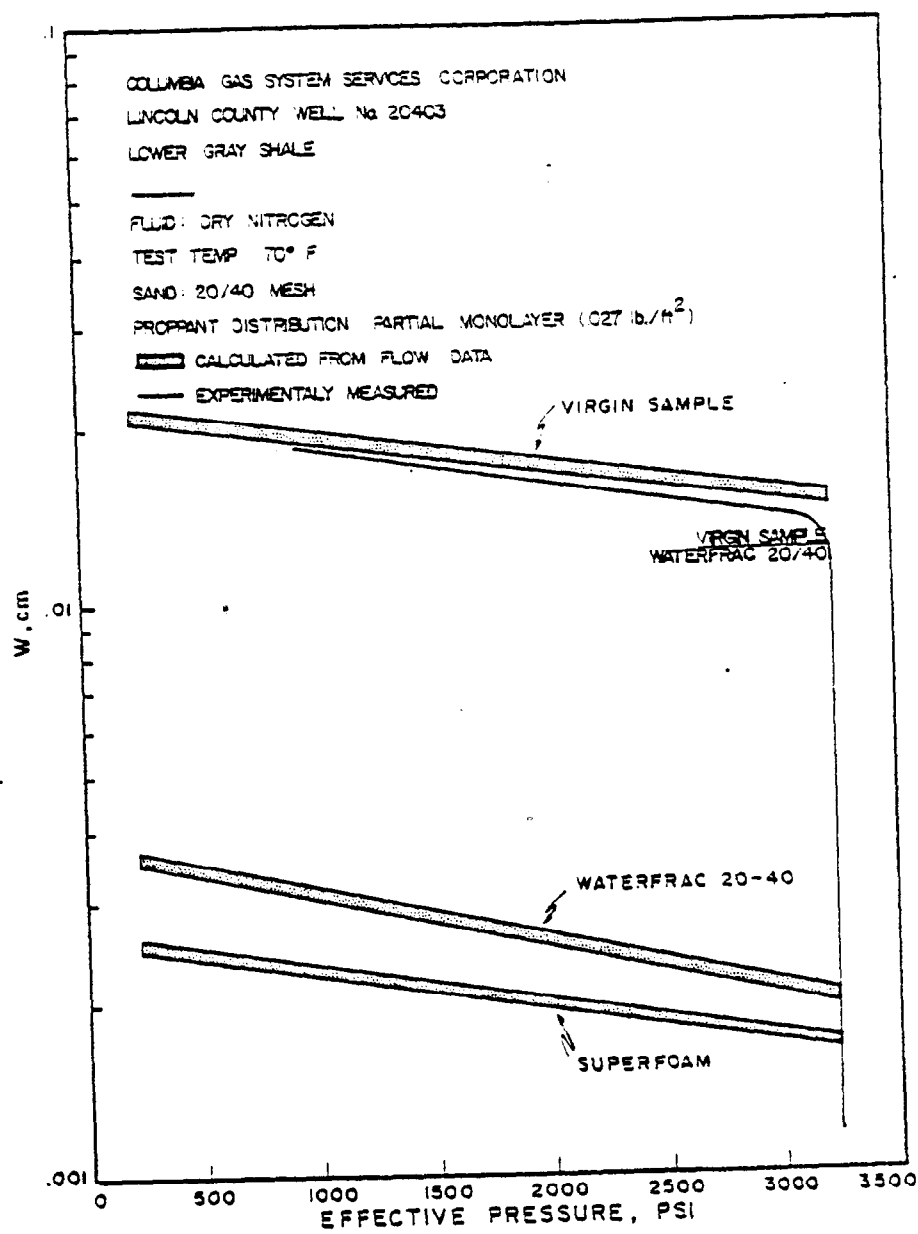


Figure 5. Trend of the effective fracture width with the increase in effective pressure for 'Lower Gray Shale'.



Figure 6. Fracture face of the 'Lower Gray Shale' sample interacted by Waterfrac 20-40.

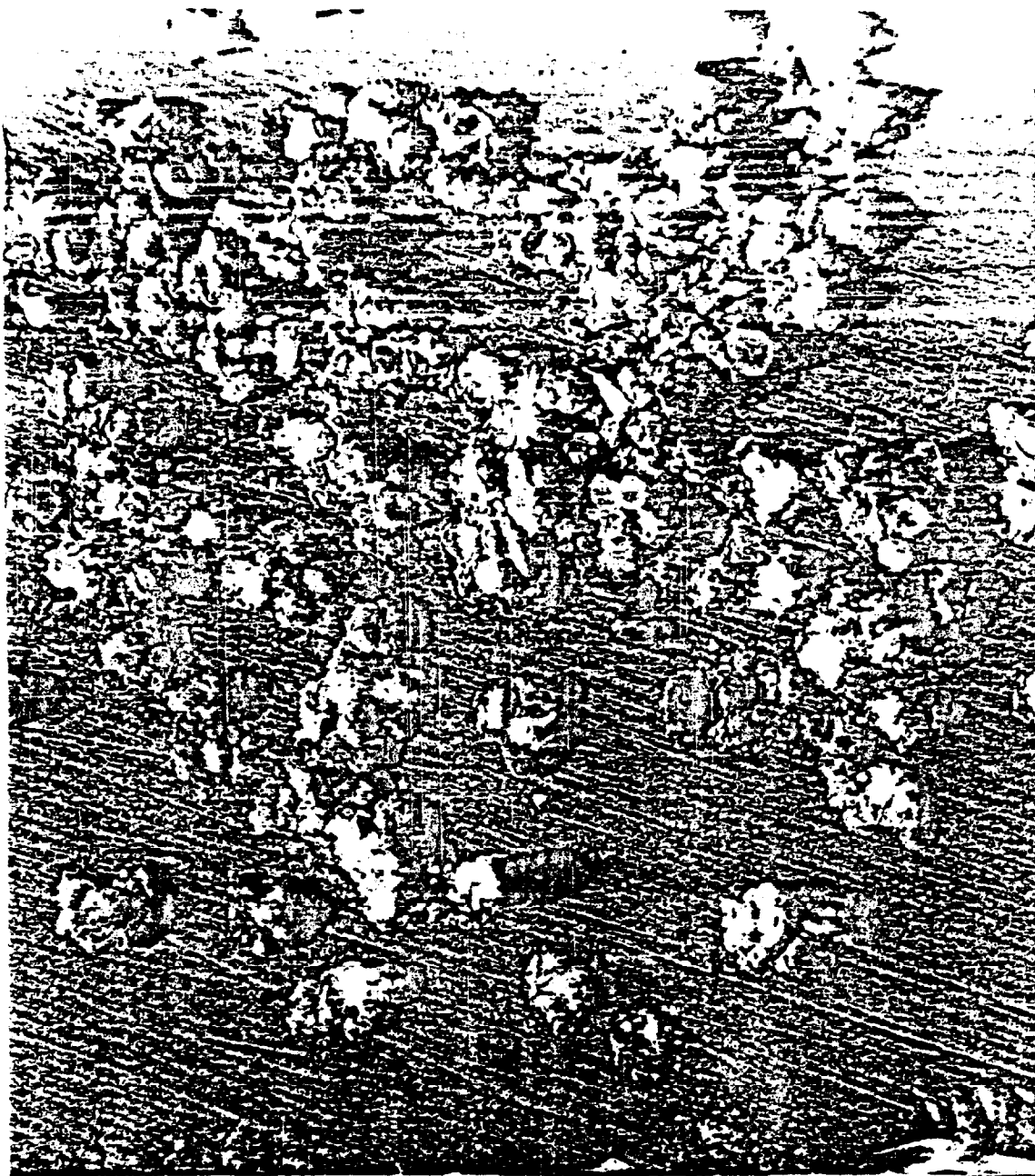


Figure 7. Fracture face of the 'Lower Gray Shale' sample interacted by Superfoam.



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# APPENDIX

## FLOW CAPACITY MEASUREMENTS

The flow capacity of a fracture (a product of fracture permeability and width of the fracture) is usually reported instead of the permeability because the fracture width is generally not known.

The calculation of the flow capacity of a fracture follows from a simple derivation of Darcy's law, presented by Amyx, *et al.*, (1960)

$$Q_0 = \frac{1 \times 10^{-3} KA(P_i - P_o)}{\mu L} \quad (1)$$

where,

$Q_0$  = flow rate of outlet fluid (ml/sec)

$K$  = permeability (millidarcy's)

$A$  = cross-sectional area of flow (cm<sup>2</sup>)

$P_i$  = inlet pressure (atm absolute)

$P_o$  = outlet pressure (atm absolute)

$\mu$  = viscosity of fluid (centipoises)

$L$  = length of the sample (cm)

For a fracture the cross-sectional area ( $A$ ) of flow is essentially:

$$A = W \times h \quad (2)$$

where,

$W$  = width of the fracture, cm

$h$  = height of the fracture, cm.

Substitution of Equation (2) into (1) results in the following relationship or the flow capacity,  $KW$ , in md-cm.

$$KW = \frac{1 \times 10^3 \cdot L \mu Q_o}{h(P_i - P_o)} \quad (3)$$

In the reported tests, nitrogen flowed in and out of the pressure vessel through small lines with resulting pressure losses; therefore, a second set of lines were used to sense gas pressures at the ends of the samples. In this way, and for steady state flow, pressures were measured directly and no corrections were needed for line losses. The gas flow rate was measured at atmospheric pressure at the end of small flow lines leading from the pressure vessel. The volumetric flow through the samples was determined by making the pressure correction between the flowmeter (at atmospheric pressure) and the sample mean pore pressure (assuming isothermal flow at 70°C). Figure 8 is a schematic diagram of the experimental set-up.

An estimate of the width of the flow channels in the unpropped fracture can be made assuming equivalent permeability for flow between parallel plates (Craft and Hawkins, 1959).

$$Q_o = \frac{W^2 A (P_i - P_o)}{1.74 \times 10^{-6} \mu L} \quad (4)$$

Here again replacing the term  $A$  by Equation (2), we have:

$$W = \left\{ \frac{1.74 \times 10^{-6} \mu L Q_o}{h(P_i - P_o)} \right\}^{1/3} \quad (5)$$

This same equation can be used to make an estimate of the *effective* width of the flow channels in propped fractures.

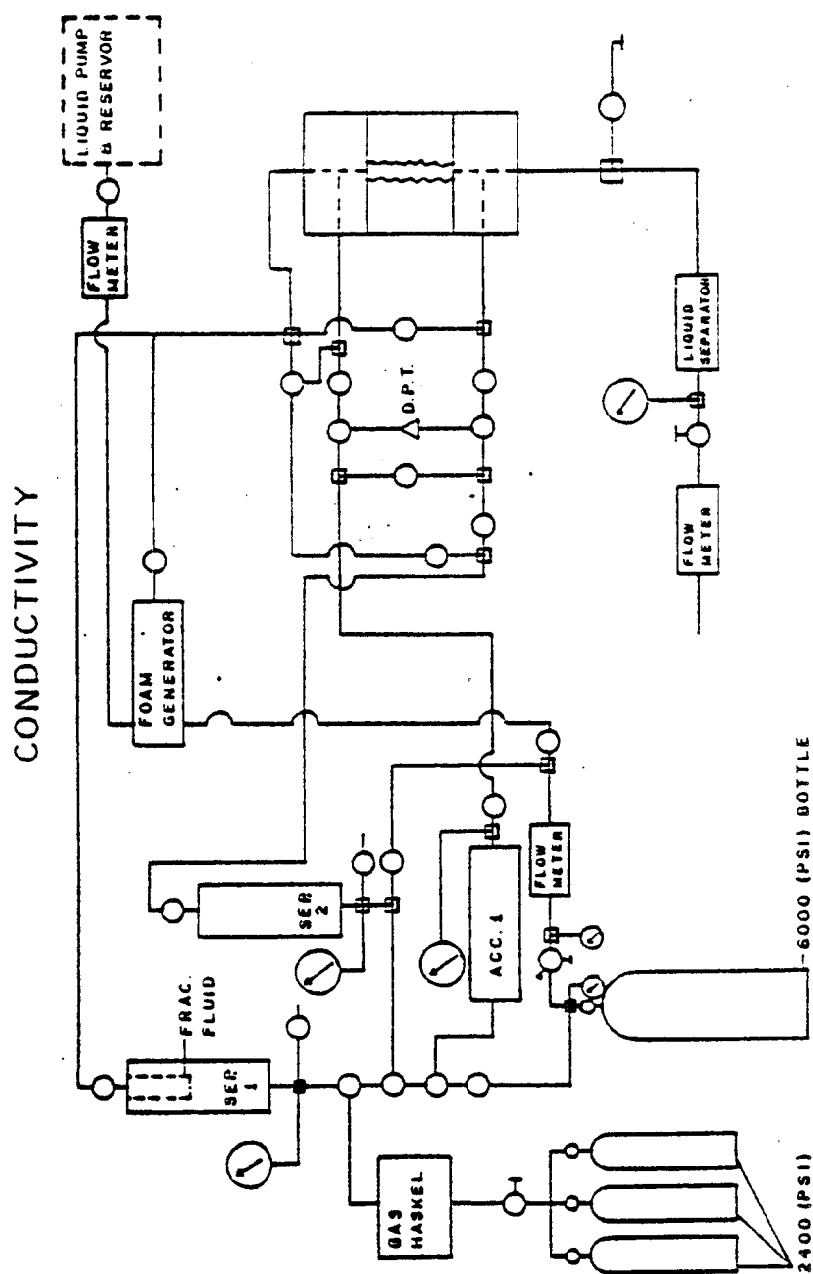


Figure 8. Schematic design of the flow set-up